GEOLOGICAL REPORT

on ·

PAUL AND MIKE CLAIMS

LEWIS CREEK AREA, WASA, B.C.

NTS 82 G/12 & G/13

FORT STEELE MINING DIVISION

Latitude 49° 46' N Longitude 115° 41.5' W

for

DIA MET MINERALS LTD.
KELOWNA, B.C.

by
PETER KLEWCHUK
May 9, 1988

Table of Contents

| | PAGE |
|--|-------------|
| ABSTRACT | 1 |
| INTRODUCTION | 2 |
| CLAIM DESCRIPTION | 3 |
| LOCATION AND ACCESS | 3 |
| PREVIOUS WORK | 5 |
| GEOLOGY REGIONAL GEOLOGY GEOLOGY OF THE PAUL - MIKE CLAIMS AREA -Fort Steele Formation -Aldridge Formation -Depositional Environment of Purcell Rocks, Kootenay Range -Structure | 6 6 7 |
| ECONOMIC GEOLOGY -Stratiform Deposits -Fissure Zone and Vein Deposits | 11 |
| MINERAL POTENTIAL OF PAUL AND MIKE CLAIMS | 17 |
| GEOPHYSICAL SURVEYS | 20 |
| GEOCHEMICAL SURVEY | 23 |
| DISCUSSION OF DRILL HOLE GEOCHEMICAL RESULTS . | 26 |
| RESULTS OF 1988 ROTARY MUD DRILLING | 27 |
| PROPOSED EXPLORATION PROGRAM | 29 |
| BOMINAMED DUDOEM OF DDOCDAM | 3.0 |

ILLUSTRATIONS

APPENDIX

REFERENCES

CERTIFICATE OF AUTHOR'S QUALIFICATIONS

ILLUSTRATIONS

| FIGURE 1 | Index Map |
|-----------|---|
| FIGURE 2 | Claim Map |
| FIGURE 3 | Cross Section from Kimberly to the east Margin of the Rocky Mountain |
| FIGURE 4 | Geophysical survey of the Paul-Mike Claims |
| FIGURE 5 | Geochemical map Heavy Minerals - Lead in soil |
| FIGURE 6 | Geochemical map Heavy Minerals - Zinc in soil |
| FIGURE 7 | Geochemical map Heavy Minerals - Copper in soil |
| FIGURE 8 | Diagrammatic Section Purcell Supergroup |
| FIGURE 9 | Lead in Soil |
| FIGURE 10 | Zinc in Soil |
| FIGURE 11 | Copper in Soil |
| FIGURE 12 | Geologic map of the Paul - Mike Claim Group |

ABSTRACT

This report describes the geology and mineral potential of the Paul - Mike claim group located east of Wasa Lake, in southeastern British Columbia. Surface geochemical surveys and rotary air and mud drilling have identified anomalous lead - silver mineralization within overburden glacial silts and clays. Geophysical surveys suggest the mineralization is located proximal to and on the west side of the regional Kootenay River Fault.

The Paul - Mike claims cover an area believed to be underlain by Proterozoic metasedimentary rocks of the Aldridge Formation which host the very large lead-zinc-silver Sullivan deposit at Kimberley about 22 kilometers to the west and the Estella and Kootenay King deposits a few kilometers to the east and southeast. The presence of a major fault near the east boundary of the claim group is compatible with bedrock of the Paul - Mike claims being near the Lower - Middle Aldridge contact zone, similar to the Sullivan orebody at Kimberley.

Anomalous lead values up to 3.66% have been obtained from rotary air drill samples collected in glacial clays on the claim block to a depth of 780 feet below surface. Geochemical results of two rotary mud drill holes which penetrated overburden to bedrock generally substantiate the presence of anomalous base metal mineralization in the overburden material but the useage of mud in drilling and the spherical nature of the lead particles in the clays was not conducive to effectively recovering the mineralization.

Rock chips recovered from the westernmost of two rotary mud drill holes and presumably from bedrock, may be of the Middle Aldridge Formation. If these chips are from bedrock and the Middle Aldridge interpretation is correct, the base metal mineralization detected in overlying glacial sediments could be from sulfide mineralization at the base of the Middle Aldridge Formation.

It is recommended that a program estimated to cost \$250,000.00 be conducted to further evaluate the distribution of the anomalous overburden minsralization and to test bedrock for the presence of a stratiform base metal deposit.

INTRODUCTION

The Paul Mike claims located just east of Wasa Lake in southeastern British Columbia (Figure 1), were staked to cover coincident soil geochemical anomalies and airborne geophysical anomalies. It is believed the anomalous mineralization reflects a bedrock source of lead-zinc-silver of the Sullivan stratiform type.

Geochemical analysis of heavy mineral concentrates collected from overburden clays and silts on the claim group with a rotary air drill in 1985 gave strongly anomalous values in lead and silver.

Two rotary mud drill holes were completed on the property in early 1988. Bedrock was encountered in the eastern hole at a depth of 915 feet; the hole was completed in bedrock to 938 feet. The western hole was drilled to 1991 feet and may have encountered bedrock at about 1940 feet. The deepest hole of the 1985 program was deepened from 780 feet to 960 feet in overburden clays and silts but technical difficulties prevented further drilling of the hole.

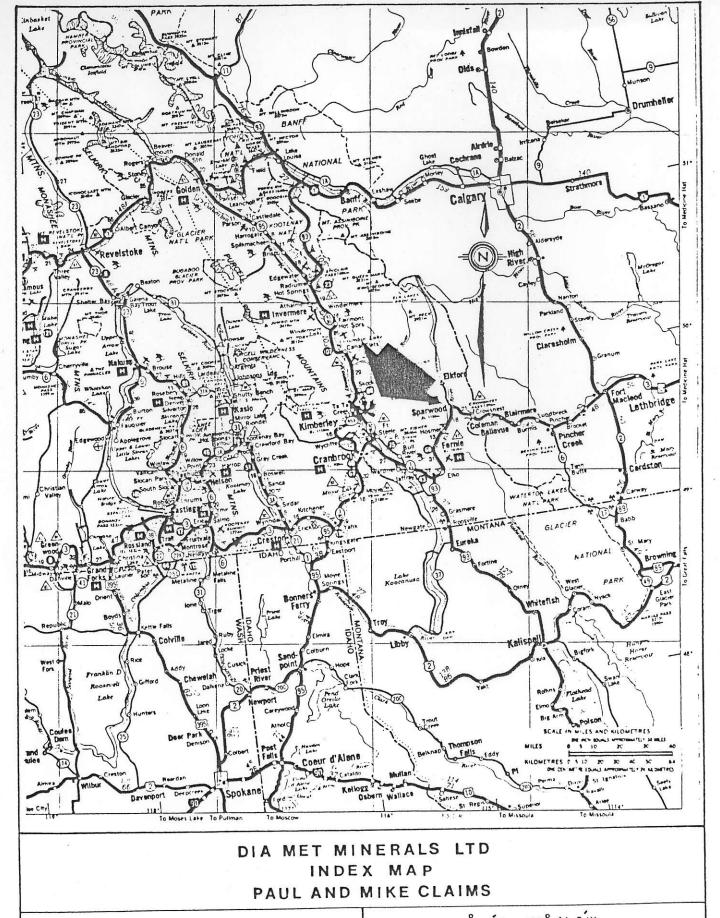
CLAIM DESCRIPTION

The Paul - Mike claim group consists of the following contiguous claims (Figure 2):

| CLAIM NAME | UNITS | RECORD NO. | RECORD DATE | EXPIRY DATE |
|---------------|-------|------------|----------------|-------------|
| Paul 1 | 18 | 1349 | Feb. 17, 1981 | 1991 |
| Paul 2 | 18 | 1350 | Feb. 17, 1981 | 1992 |
| Paul 3 | 18 | 1351 | Feb. 17, 1981 | 1994 |
| Mike 3 | 3 | 1648 | Aug. 9, 1982 | 1990 |
| Mike 4 | 6 | 1649 | Aug. 9, 1982 | 1992 |
| Mike 5 | 18 | 1650 | Aug. 9, 1982 | 1990 |
| Mike 6 | 2 | 1651 | Aug. 9, 1982 | 1992 |
| Mike 7 | | 2222 | Aug. 22,1984 | 1990 |
| Mikey No.1 Fr | 1 | 1772 | April 26, 1983 | 1995 |
| Alexis | 12 | 2922 | June 10,1987 | 1989 |
| Jimmy | 12 | 2923 | June 10,1987 | 1989 |
| Alexis 1 | 16 | 2924 | June 10,1987 | 1989 |
| Jimmy 1 | 18 | 2925 | June 10,1987 | 1989 |

LOCATION AND ACCESS

The Paul - Mike claim group is located on the eastern margin of the Rocky Mountain Trench, at the foot of the west-facing slope of the Hughes Range, immediately east and south of Wasa Lake. Elevations on the property range from 800 to 1100 meters. Wasa Lake at the northwest corner of the claim group is 20 kilometers east of Kimberley at latitude 49° 46'N and longitude 115° 41', on NTS map sheets 82 G/12 and G/13. Access to the claims is provided by highway 93/95, the Lewis Creek road and numerous logging roads and bush roads.



B2G/12E & B2G/13E

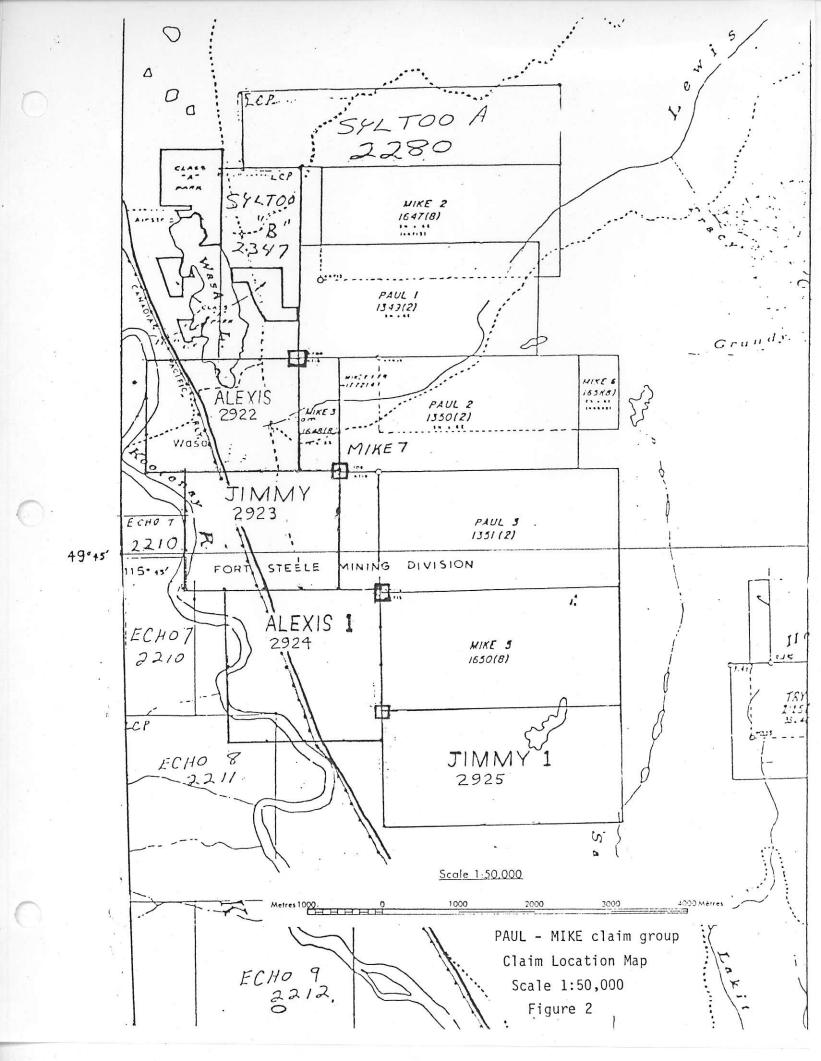
49° 46'N 115° 41.5'W

GOWER; THOMPSON & ASSOCIATES

Drawn J. F. B.

K.E. NORTHCOTE AND ASSOCIATES LTD

April 30 1983



A geological report for Dia Met Minerals Ltd. on the Paul - Mike claims by E.A.Schiller dated September, 1987 provides a summary of the exploration history, geology and mineral potential of the property. Pages 5 to 26 immediately following are from the Schiller report.

PREVIOUS WORK

Prior to the staking of the claims in 1982/83 there was no recorded work on the claims. Based on a geological model in which projections of stratigraphy eastward from Cominco's Sullivan lead-zinc mine, west of Kimberley, the Wasa area was subject to geological, geochemical and geophysical investigation in 1982 - 85. Utilizing information from Hoy (1978) it was postulated by Fipke (unpublished report) that the Proterozoic Middle Aldridge Formation could extend eastward beneath Wasa Lake and be possibly mineralized. The presence of the Kootenay King and Estella lead-zinc mines east of Wasa Lake supported this hypothesis (Figure 3).

In 1982, an airborne electromagnetic survey (DIGHEM) was completed over the area and a resistivity anomaly was detected on the eastern edge of the claims (Figure 4). A weak electromagnetic conductor was mapped in the south east part of the claims. Due to the highly conductive nature of the overburden gravels estimated to be 100 to 110 metres by DVORAK of Dighem (personal comm.) but found to be over 700 feet in the rotary drilling, the interpretation of the geophysical data must be considered inconclusive.

In 1985 Induced Polarization surveys were conducted over the claims and two anomalous areas were delineated that coincided with geochemical anomalies for lead, zinc and copper (Figures 5, 6, & 7). Dia Met Minerals Ltd. optioned the claims to Bearcat Exploration Ltd. of Calgary, Alberta who drilled, in 1985, three rotary holes totalling 546 metres (Appendix).

In 1986, Dia Met prepared heavy mineral concentrates from 10 foot intervals over the interval 600' - 780' from hole No. 3. The concentrates were analyzed for 30 elements and appendixed in this report. Lead and silver were significantly anomalous whereas the zinc content was only moderatly anomalous in two, ten foot intervals. Note worthy is the consistent elevated levels of lead exceeding 10,000 ppm (1.0%).

GEOLOGY*

REGIONAL GEOLOGY

The Paul and Mike claims area is underlain by clastic rocks of the Purcell Supergroup and lie within the Fernie (West Half) map sheet (Leech, 1966) and on the west margin of the Estella-Kootenay King area (Hoy, 1979). It is from these maps, and papers by T. Hoy, 1982: M.E. McMechan, 1981: M.E. McMechan, T.Hoy and R.A. Price, 1980, that the following account of the regional geology is summarized.

The Purcell clastic rocks exhibit marked changes of facies from east to west reflecting different environments of deposition. In general these rocks form a series of sedimentary wedges that thicken and become more argillaceous westwards. To the east, in the Clark Range, most of the Purcell rocks of the Waterton, Altyn Formations are shallow-water shelf, tidal flat flood-plain or deltaic deposits. This thin platform thickens to the west and in the Purcell Mountains becomes more argillaceous having the same stratigraphic position as the thick sucession of dark laminated argillite, siltstone, sandstone, turbidite deposits of the Aldridge Formation. With continued subsidence the basin of turbidite deposition expanded. In the Clarke Range overlying clastic facies of Appekunny and Grinnell Formations thicken westwards correlative with the Creston Formation. Similarly the overlying Kitchener-Siyeh Formation thickens and becomes more argillaceous to the west. Purcell sedimentary rocks, overlying the Purcell Lava on the east, form thin lithologically distinct formations of tidal flat or floodplain deposits while correlative rocks in the Purcell Mountains to the west are thicker, contacts

^{*} The fellowing description (pages 6 - 25) was taken from a report written by Northcote and Gower and Thompson in June, 1983 for Dia Met Minerals Ltd.

are more gradational and dark argillaceous facies are more prominent.

Hoy postulates that syndepositional transverse faults, the Moyie-Dibble Creek and St. Mary-Boulder Creek faults, originating in Purcell time locally modified the depositional pattern of Purcell rocks. These faults cross the Rocky Mountain Trench south of the Kootenay King area and affected deposition of Purcell rocks in this area. Fort Steele coarse fluvial quartzites at the base of the succession may record an initial arching of the crust prior to development of a cross cutting fault controlled structural basin to the south. Thick deposits of turbidite and wacke beds were deposited in the basin in aldridge time while generally more shallow-water sediments accumulated in the Kootenay King area to The Boulder Creek fault, a right lateral reverse fault, is located along the approximate site of the Lower to Middle Aldridge growth fault with its associated dramatic thickness and facies changes. North of the Kootenay King area in the direction of current transport and deepwater environment the quartzite and surrounding strata thin appreciably and become finer grained, suggesting deposition in a small fault bounded structural depression. Overlying the Fort Steele Formations are coarse grained channel deposits in the inner part of a submarine fan consisting of quartzite and locally conglomerate in calcareous siltstone and argillite which were probably a source of turbidite deposits in correlative Middle Aldridge rocks to the north, south and west.

THE GEOLOGY OF THE PAUL AND MIKE CLAIMS AREA

1 ----

The Paul and Mike claims are largely underlain by unconsolidated material which covers the probable position of the Kootenay River fault which follows much of the eastern side of the Rocky Mountain Trench in the Fernie (West Half) map sheet.

The Fort Steele Formation crops out on the east edge of the claims area and has moderate westerly dip towards the centre of the claims. Here, the Fort Steele Formation represents the upper limb of a recumbent fold whose axial plane strikes north-north-westerly and dips westerly. The Fort Steele Formation begins dipping easterly at about 1050 metres elevation east of Sowerby Lake and from there proceeds upwards in section to 2000 metres elevation where Aldridge Formation overlies Fort Steele conformably in the south part of the map area and along a fault which transects bedding at a low angle from a point east of Sowerby Lake northerly to Lewis Creek. Successively younger members of the Aldridge Formation are exposed eastwards to the Aldridge-Creston contact two kilometers east of Estella mine.

Fort Steele Formation*

The total thickness of the exposed Fort Steele section is greater than 2000 metres with the base not exposed. The formation comprises at least three generally upward fining sequences, or megacycles, several hundred metres thick. The megacycles grade, with some reversals, from coarse massive or cross bedded quartzites at the base to laterally more persistent thinly laminated siltstone and argillites at the top.

Quartzites at the base of the megacycles are medium to coarse grained forming discontinuous beds up to a metre thick which thin and die out laterally. They are commonly structureless or crudely layered and scour the underlying unit producing broad troughs. Up section, within each of the megacycles, quartzites are finer grained, more argillaceous, thinner bedded and more persistent laterally. The relative proportion of the siltstone/argillite component increases at the top of upward fining sequences. Siltstones at the top of the megacycles are thin and horizontally

^{*} as described by Hoy, 1979, 1982

laminated or ripple cross laminated. Individual beds grade up to dark laminated argillite that contains abundant desiccation cracks. Lenticular bedding and silt scours are common. At the top of the Fort Steele Formation the quartzite/siltstone component gradually decreases grading into Aldridge Formation; which boundary is placed above the last occurrences of cross bedded quartzite or observed desiccation cracks in argillite.

Aldridge Formation

Rusty weathering argillite, siltstone and shale of the Aldridge Formation conformably overlie the Fort Steele Formation and is divided into three main divisions. The lowest division, A 1, is dominantly laminated and ripple cross-laminated siltstone, finely laminated dark argillite, quartzite and, near the base, a dolomite member containing intercalated chert. Quartzite beds that contain basal rip-up clasts and occasional northerly trending flute casts are interlayered with a buff-weathering, commonly graded and lenticular-bedded siltstone near the middle of A 1.

The middle division of the Aldridge Formation, A 2, is approximately metres in thickness and consists of interlayered siltstone, quartzite beds separated by dark rusty weathering argillite. The quartzite beds are generally massive, although the top few centmetres may be crudely layered and finer grained than the base. Flute casts on the base of some quartzite beds indicate northward transport.

The upper division, A 3, comprises 500 to 800 metres of finely laminated, dark argillite and siltstone. It is transitional into green quartzite and siltstone at the base of the overlying Creston Formation.

Depositional Environment of Purcell Rocks, Kootenay Range

The following account of depositional environment of Purcell rocks in the Kootenay King-Estella area is taken from Hoy, 1979.

"As in the Clarke Range, the bulk of the Purcell rocks in the Kootenay Range record deposition in a shallow water tidal flat or flood plain environment. However the lithology of the Aldridge Formation indicates that it was deposited in deep water."

"The Fort Steele Formation at the base of the Purcell Supergroup is unique. It consists predominantly of braided fluvial deposits derived from a source area to the south (Hoy, 1978). A marine transgression is apparent in lower Aldridge time, and the alluvial fan deposits of the Fort Steele are overlapped by intertidal and subtidal mud flat deposits, which gave way upward to slightly crenulated and laminated carbonates that are similar to subtidal algal mat deposits. Overlying laminated siltstone and argillite represent continuing transgression. In the Kootenay King area, the lower division of the Aldridge Formation, (Al), thins from approximately 2500 metres thickness in the south to less than 1500 metres only 15 kilometres to the north. In addition, the character of lower division rocks also changes northward. Thick sections of quartz-rich siltstone in the south give way to finer grained and more dolomitic siltstone to the north and finally grade laminated black argillite. As well, thick coarse grained quartzites that contain zones of small angular clasts near their base become thinner and finer grained northward. The thinning and fining of sediments to the north, reinforced by a northerly current transport direction that is indicated by sedimentary structures, suggests that a local fault-bounded basin, with south side up may have developed in lower Aldridge time. The Boulder Creek fault, an eastern extension of the St. Mary Fault which Lis and Price (1976) have shown to be active in Hadrynian time, appears to have been the focus of an Helikian fault that defined the southern limit of the basin. The coarse grained guartzite layers are interpreted to be channel deposits that cut silty levee or back-levee beds in the inner part of a submarine fan. Movements on the fault might have triggered deposition of the channel sand deposits. Kootenay King, a stratiform Pb-Zn deposit, is in the upper part of the thickest and coarsest of these quartzite sequences. Quartzite interlayered with conglomerate just north of the transverse Lewis Creek-Nicol Creek fault zone... also thins, becomes finer grained and dies out to the north suggesting that this zone also was active in lower Aldridge time and locally controlled sedimentation."

"The middle division of the Aldridge Formation (A2) consists of proximal turbidite deposits, transported in a northerly direction. The first evidence of a marine regression appears near the top of the Aldridge Formation (Hoy, 1978) and green siltstone and shale near the base of the overlying Creston Formation were deposited in a subtidal environment. Overlying purple siltstones that contain numerous mud cracks, rip-up clasts and ripple marks are tidal flat or flood plain deposits similar to those described in the Clark Range. The carbonate rocks of the Kitchener-Siyeh, and the mud-cracked siltstone and shale of the Gateway, Phillips and Rooseville Formations, were also formed on extensive tidal flats or flood plains."

Structure

The structure of the Purcell rocks in the Estella-Kootenay King map area is dominated by a large, open, recumbent anticline. Its axial plane dips to the west and, throughout most of the western part of the map area, bedding in its upper limb dips to the east. It is noted, however, on field work sheets provided by Hoy, that outcrops on the east margin of the Paul and Mike claims have consistent gentle to moderate westerly dips to a point of flexure east of Sowerby Lake. Because of this westward dip of the upper limb of the recumbent fold in this local area it is possible that Aldridge Formation higher on this limb could subcrop in the claims area. If the projection of the Kootenay River fault passes through the claims area it would transect and displace the upper limb of the recumbent fold. Because the direction and magnitude of movement of opposite sides of the Kootenay River Fault are not known it is not possible to predict with certainty what formation underlies the overburden on the west half of the Paul-Mike claims.

ECONOMIC GEOLOGY

Mineral deposits in rocks of the Purcell Supergroup are of two major types:

- a) the stratiform type represented by Sullivan,
 North Star, Stemwinder, Vulcan and Kootenay King
 and
- b) vein or fissure filling or replacement type represented by St. Eugene, Society Girl, Aurora and Estella deposits.

STRATIFORM DEPOSITS

Stratified Pb, Zn, Ag (Cu, Au)

Sullivan

The Sullivan mine, at Kimberley, B.C., one of the largest base metal deposits in the world, is a stratiform Pb, Zn, Ag, Cd, Sn, Fe deposit within rocks of the Purcell Supergroup at the Lower-Middle Aldridge boundary. The deposit is considered to be a hydrothermal, synsedimentary deposit that formed in a small submarine basin. The Sullivan mine has produced more than 100 million tonnes of ore and has reserves of approximately an additional 50 million tonnes grading 4.9% Pb, 6.1% Zn and 37 g/tonne Ag.

The ore deposit is zoned laterally and vertically. The western part is massive pyrrhotite with wispy layers of galena overlain by layered pyrrhotite, sphalerite, galena and pyrite intercalated with clastic beds. The eastern part has several distinct conformable layers of generally well-laminated sulphides separated by clastic rock. At the outer limits of the deposit are iron sulphide bands.

The footwall of the massive western section has north trending brecciated conduit zones and is altered to dark chert-like tourmaline-rich rock. Albite-chlorite-pyrite alteration is also restricted to the western part of the orebody. The deposit is zoned with Pb, Zn and Ag values decreasing towards the eastern margin. Tin is concentrated in the western part. Higher absolute metal values and higher Pb/Zn and Ag/Pb ratios overlie breccia zones.

Kootenay King

The Kootenay King mine is located on the north side of the Wildhorse River, 10 miles by road from Fort Steele. This is a small lead-zinc stratiform deposit of less than 100,000 tonnes in Middle Aldridge clastic sedimentary rocks.

Mineralization occurs near the hinge of a large anticlinal fold that is thrust eastward onto younger Paleozoic strata. It consits of a layer of finely laminated sphalerite, galena and pyrite intercalcated with dolomitic to argillaceous siltstone in Kootenay King quartzite. Near its base the quartzite contains coarse angular lithic clasts interpreted as a channel sandstone deposit similar to both overlying and underlying siltstone.

Over a two year period, 1952 and 1953, the mine produced 14,610 tonnes of ore yielding 0.715 kg of Au, 0.90 tonnes of Ag, 710 tonnes of Pb and 881 tonnes of Zn.

Possible Controls for Stratiform Mineralization

A number of stratiform lead-zinc deposits are preferentially located in the vicinity of synsedimentary transverse fault zones coincident with the St. Mary-Boulder Creek and Moyie-Dibble Creek

transverse faults south of Kimberley and the Hall Lake fault to the north. These deposits are related to areas of tour-malinization and intraformational conglomerates and breccias which are associated with synsedimentary faults. This suggests a genetic link between mineral deposits, small basins and related fault systems. Local thickening of host succession suggests further that third-order basins cross cutting a regional north-east trending second order rift structure may have been an important local control. (Hoy, 1982)

Stratiform Copper Potential

There are numerous widespread occurrences of stratiform copper in Purcell rocks in Southeastern B.C. These occurrences are in fluvial or shallow water deposits most notably in the Grinnell-Creston Formations and to a lesser extent in the Kitchener-Siyeh, Gateway, Phillips, Kintla and Rooseville Formations in the Clarke Range and on the east side of the Rocky Mountain Trench. There is evidence of some enrichment or remobilization and redeposition of copper in association with Purcell sills and dykes. There has not, however, been any stratiform copper occurrences found to date with production potential in southeastern B.C.

Similar deposits with production capability are located in Grinnell-Creston equivalent rocks in Glacier Park Montana and a notable deposit reported to be producing at a rate of 2.72 million tonnes per year with reserves greater than 55 million tonnes 0.74% Cu and 48 grams Ag per tonne in Rivette quartzite at Troy, Montana, (Hoy, personal comm.).

In the Estella-Kootenay King area there are a number of stratiform copper occurrences in Fort Steele and Aldridge

quartzite, none of which have indicated significant reserve potential. See Table I.

Fissure Zone and Vein Deposits

There are a number of Pb, Zn, Ag fissure zone replacement and vein deposits within rocks of the Purcell Supergroup. These include the St. Eugene-Society Girl and Aurora deposits on the east and west sides respectively of Moyie Lake 25 km southwest of Cranbrook. The estella mine is also included in this group and is located 5 km east of the Paul and Mike claims.

St. Eugene-Society Girl

The St. Eugene mine is in Upper Aldridge Formation and occurs between two main fissure zones striking east west and dipping 70°S. The ore forms replacement deposits in quartzite and is restricted to the fractured area between 2 main fissures. The veins are narrow with mineralization consisting of galena, sphalerite, magnetite and minor chalcopyrite. This mine has produced a total of 1,475,266 tonnes of ore yielding 78.8 kg gold; 182,690 kg Ag; 113,035 tonnes Pb and 14,483 tonnes Zn. This mine ceased operations in 1929.

The Society Girl is at the Upper Aldridge-Creston Formation contact and is in a narrow vein cutting thin bedded argillaceous quartzites. The vein is oxidized containing cerrusite, pyromorphite, galena and sphalerite with little or no gangue. The Society Girl produced 2,984 tonnes of ore yielding 432 kg Ag, 580 tonnes of lead and 24 tonnes zinc. The Society Girl mine ceased operations in 1952.

The Estella mine is located in the headwaters of Tracy Creek about 8 kilometres in a straight line from Wasa or 26 kilometres by road. The mill was located at Wasa and the ore was hauled by truck to the mill. This possible source of contaminants should be borne in mind when interpreting geochemical results in the area adjacent to the haulage road.

The mineralized rocks at the Estella mine are northwest striking, southwest dipping argillites and quartzite of the Aldridge Formation. Small intrusive bodies are also present; a syenite exposed near the portal and a sill-like diorite which also contains mineralization.

The lode is a zone of fracturing and light shearing and is semi-bedded in the sedimentary rocks and penetrates the diorite. The ore is a replacement by sphalerite, galena and pyrite, accompanied by more or less silica. Vein guartz is not abundant except in diagonal veins which in general contain little if any sulphides.

Intermittent production from 1951 to 1967 from the Estella mine totalled 109,518 tonnes and yielded 2.05 kg Au, 6.3 tonnes Ag, 5,181 tonnes Pb, 9,834 tonnes Zn and 1.2 tonnes Cu.

Table I summarizes the mineral properties in the general area of the Paul and Mike claims.

TABLE I

| BC . MINFILE NO. | NAME | COMMODITIES | FORMATION | TYPE OF DEPOSIT | PRODUC- TION |
|------------------------|---------------------|-----------------------|-----------------------------|---|-----------------|
| 82GNW008 | Estella Mine | Zn,Pb,Ag, Au,Cd,Cu | Aldridge & Intrusive | Shear Vein Replacement | Yes |
| 82GNW009 | Kootenay King | Ag,Pb,Zn Cd,Au,Cu | Aldridge | Stratiform | Yes |
| 82GNW016 | Kootenay Selkirk | Cu,Pb,Ag | Aldridge Purcell Sill | Purcell Sill Fractures, Replacement | · - |
| 82GNW017 | Try Again | Cu | Fort Steele | Fault Replacement | |

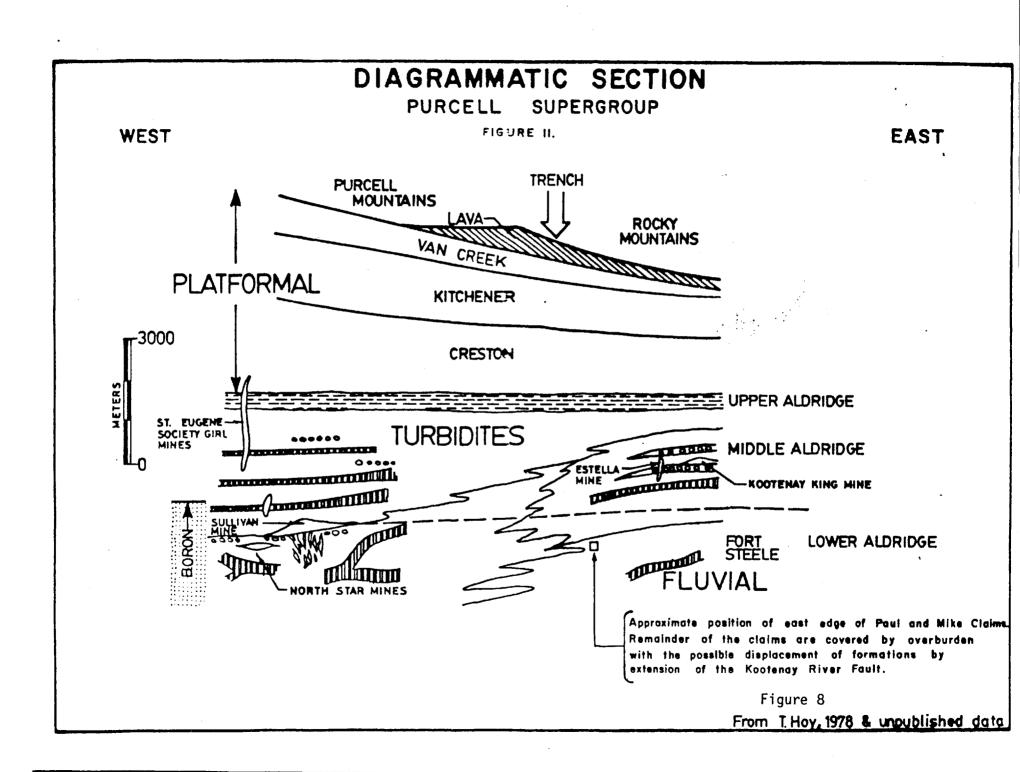
| 82GNW035 | Golden Fleece | Cu | Fort Steele | Vein | - |
|----------|------------------|------------|-------------------------|--------------------------|-----|
| 82GNW045 | Emily Tiger | Au, Ag, Pb | Fort Steele | Vein | _ |
| 82GNW046 | Wanda B | Au, Ag, Pb | Ft.Steele ? Aldridge | Vein | · |
| 82GNW059 | Lazy 19 | Cu | Ft.Steele | Vein Strata- bound | ••• |
| 82GNW061 | Lazy 32 | Cu | Ft.Steele Intrusive | Vein Fault | - |
| 82GNW062 | Cher | Cu | Ft.Steele | Strata- bound | - |

Of the above properties the Estella and the Kootenay King are the most significant.

MINERAL POTENTIAL OF PAUL AND MIKE CLAIMS

Stratiform Deposits

For comparative purposes, diagram Figure 8 summarizes Purcell Supergroup stratigraphy, environment of deposition and indicates the position of significant deposits within the sequence. This diagram indicates that the environment of the PAUL and MIKE claims-Estella-Kootenay King area with its fluvial to fanglomerate sediments differs from the deep basinal and turbidite environment of the North Star and Sullivan deposits. Clastic hosted stratiform Pb, Zn deposits have, so far, been found in Lower and Middle Aldridge sediments. The Vulcan and Sullivan deposits occur at the Lower-Middle Aldridge boundary. North Star and Stemwinder deposits just south of Sullivan are in Lower Aldridge siltstone. The Kootenay King mine, east of the PAUL-MIKE claims, is in Middle Aldridge siltstone.



The Paul and Mike claims are in a favourable position relative to the Lewis Creek and Boulder-Dibble Creek synsed-imentary transverse faults which appear to be related to stratiform Pb, Zn, Ag deposits. The presence of the Kootenay King stratiform deposit supports the contention that additional stratiform Pb, Zn, Ag deposits of unknown size may be found in similar environments in this area. There is uncertainty regarding presence, sense of movement and age of earliest displacement on the Kootenay River Fault. Suggestions of what formations may underlie the overburden on the west side of the Paul and Mike claims and their environment of deposition are speculative. It is possible that formations younger than Fort Steele underlie the overburden there and that these rocks may be favourable hosts for stratiform Pb, Zn, Ag stratiform deposits of unknown size.

Replacement Deposits

Deep seated structures other than synsedimentary, structure and environment of deposition have produced significant Pb, Zn, Ag (Au) replacement deposits in rocks of the Purcell Supergroup. For example, fissure-vein replacement deposits of Moyie Lake area, and in particular the St. Eugene mine which produced 1,475,266 tonnes of ore, were dependent upon deep seated fracture systems to provide permeability for metal-bearing hydrothermal solutions and for localization of mineralization.

The Paul and Mike claims are crossed by the synsedimentary Lewis Creek Fault and probably by the extension of the Kootenay River Fault, of uncertain age span, along the east margin of the Rocky Mountain Trench. The junction of these two, deep, major fault systems and subsidiary fractures provides optimum potential for circulation of metal charged hydrothermal solutions and deposition of fissure-vein replacement deposits. The Estella mine

is in this structural environment and proves potential for hydrothermal fissure-vein replacement deposits in this area.

GEOPHYSICAL SURVEYS

DIGHEM SURVEY

Dighem resistivity, electromagnetic and magnetic surveys were flown over the Paul and Mike claims area.

Resistivity Survey

There is a major north-northwesterly trending break in resistivity from approximately 133 - 177 ohm metres to greater than 1000. This break in resistivity swings northeasterly in the vicinity of Lewis Creek (Figure 4). The high resistivity values occur along the eastern edge of the survey block just to the west of outcrops of Fort Steele and Aldridge exposure to the east. The western half of the claim block is characterized by low resistivity (conductive) values 75 to 150 ohm metres.

Magnetic Survey

A magnetic anomaly trends north-northwesterly roughly following the boundary between the highly resistive and conductive zones at the eastern side of the Paul and Mike claims. A second smaller northerly trending magnetic anomaly lies on the west side of the Sowerby Lake east of the claim block (Figure 4).

Electromagnetic Survey

A weakly conductive electromagnetic anomaly trends northnorthwesterly in a resistivity trough in the southeast corner of the claim block (Figure 4).

Results

The marked break in resistivity corresponds to the projection of the Kootenay River fault through the Paul and Mike claims. A change in trend of the break in resistivity in the vicinity of Lewis Creek may be a result of the Lewis Creek fault.

Geophysical interpretive information supplied by C.Fipke of Dia Met Minerals Ltd. attributed to Z. Dvorak of Dighem, Falconbridge and Superior Oil geophysicists is as follows:

"The only magnetic fracture of possible significance is an elongated enhanced magnetic anomaly of north-north west direction in the northeastern part of the survey block. It roughly follows the boundary between the highly resistive and highly conductive zones mentioned above." (C.Fipke personal communication)

It should be noted that the presence of Purcell sills and dykes, which are more magnetic than surrounding sediments, may result in magnetic anomalies. It is to be expected that detailed ground magnetic surveys would detect more of these sills and dykes as well as anomalies resulting from mineralization.

The conductive material underlying the main part of the valley under Kootenay River is attributed to conductive river sediments. The highly resistive material east of the Paul-Mike claims is attributed to bedrock. In the claims area, " the rest of the conductive material away from the river valley occurs definitely at depth. It is speculated that the resistive and conductive zones reflect different rock formations...." (C.Fipke personal communication)

"Dr. Z.Dvorak, Dighem Ltd. geophysicist, calculated the maximum depth to the highly conductive rock formation adjacent to the enhanced magnetic anomaly to be 100 to 110 metres" (C.Fipke, personal communication)

Fipke suggest that the weakly conductive electromagnetic anomaly trending north-northwesterly in a resistivity trough in the southeast corner of the claim block is compatable with Sullivan type stratiform mineralization because galena and sphalerite are poor electromagnetic conductors (C.Fipke, personal communication)

The possible effects of high concentration of pyrite and pyrrhotite on electromagnetic conductivity at Sullivan were not discussed.

The northern extention of the conductor at the southeast corner of the claim block approximately coincides with anomalous lead geochemistry and the south end of an elongated northwest trending magnetic high feature (Figure 4). These near coincident anomalies may represent sulphide mineralization at a depth from near surface down to Dvorak's calculated maximum depth of 100 to 110 metres.

GEOCHEMICAL SURVEYS

HEAVY MEDIA SURVEYS

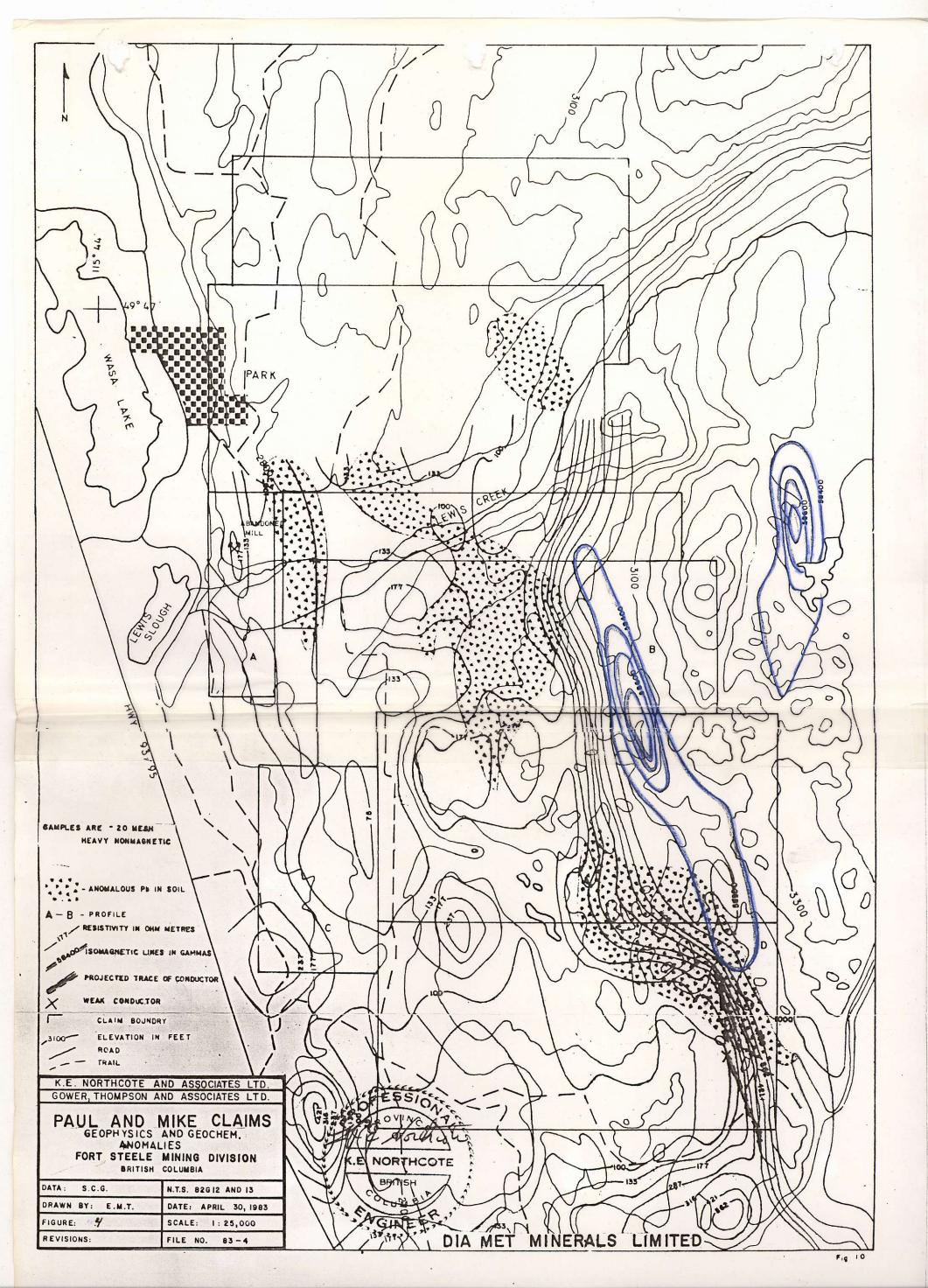
Significant anomalous concentrations of copper, zinc and lead occur in heavy media soil samples taken from glacial drift within Paul and Mike claims (Figure 2). Methodology for collection and concentration of samples is outlined in Appendix A. Concentrates of appropriate size, specific gravity and magnetic susceptability to give optimum analytical results were analyzed for Cu, Zn, Ag, Cd, As and some for Mo, Mn and Co.

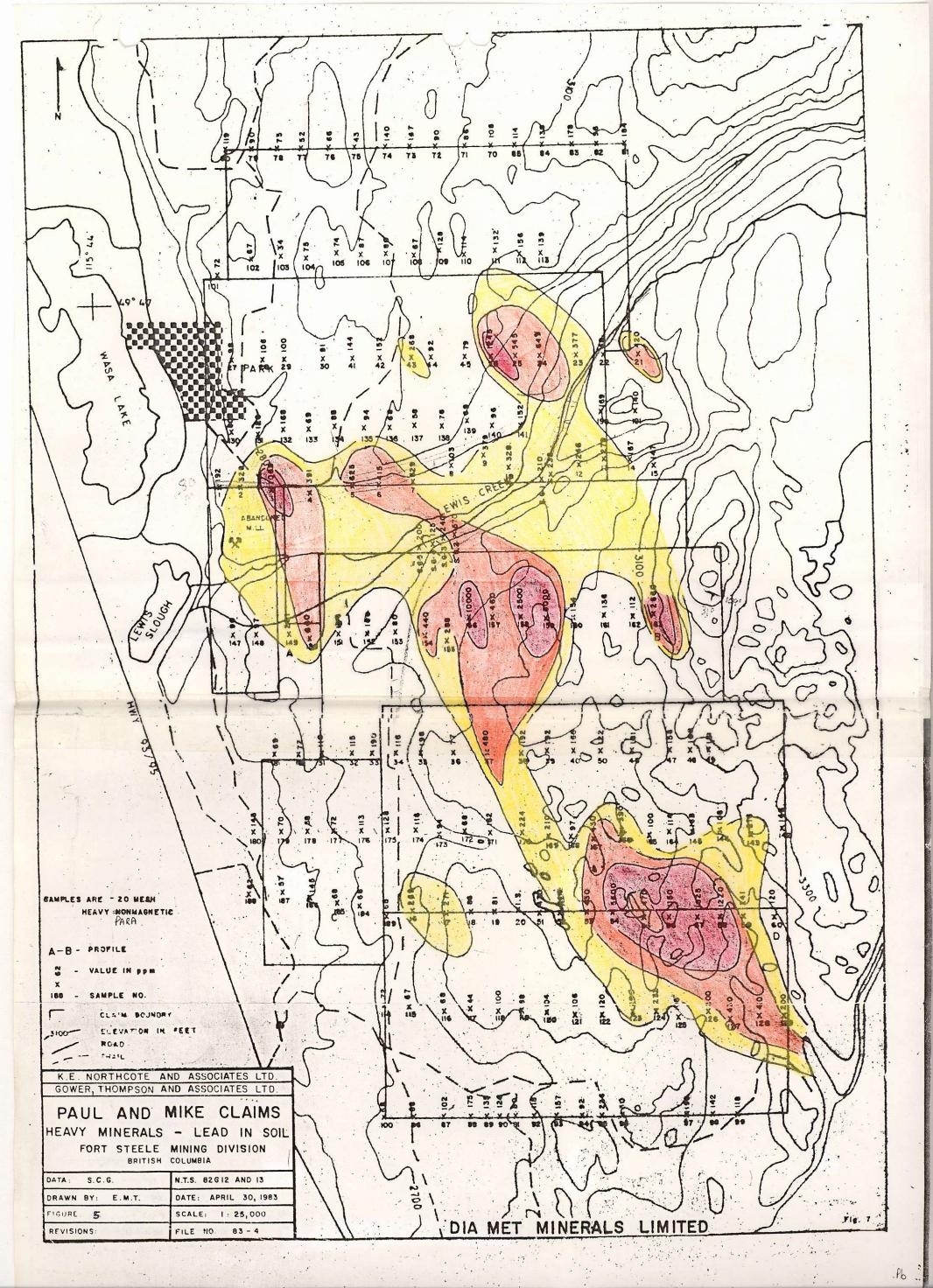
Gower and Northcote collected five samples as a procedural check near the Lewis Creek road. Care was taken to prevent any possible heavy mineral contamination resulting from spillage from trucks hauling ore from Estella mine to the mill at Wasa. Heavy media concentrates were made of these samples at C.F.. Mineral Research Ltd., Kelowna, B.C. in the same manner as outlined in Appendix A and were shipped to Bondar Clegg, North Vancouver, for Cu, Pb, Zn, Au and Ba analyses.

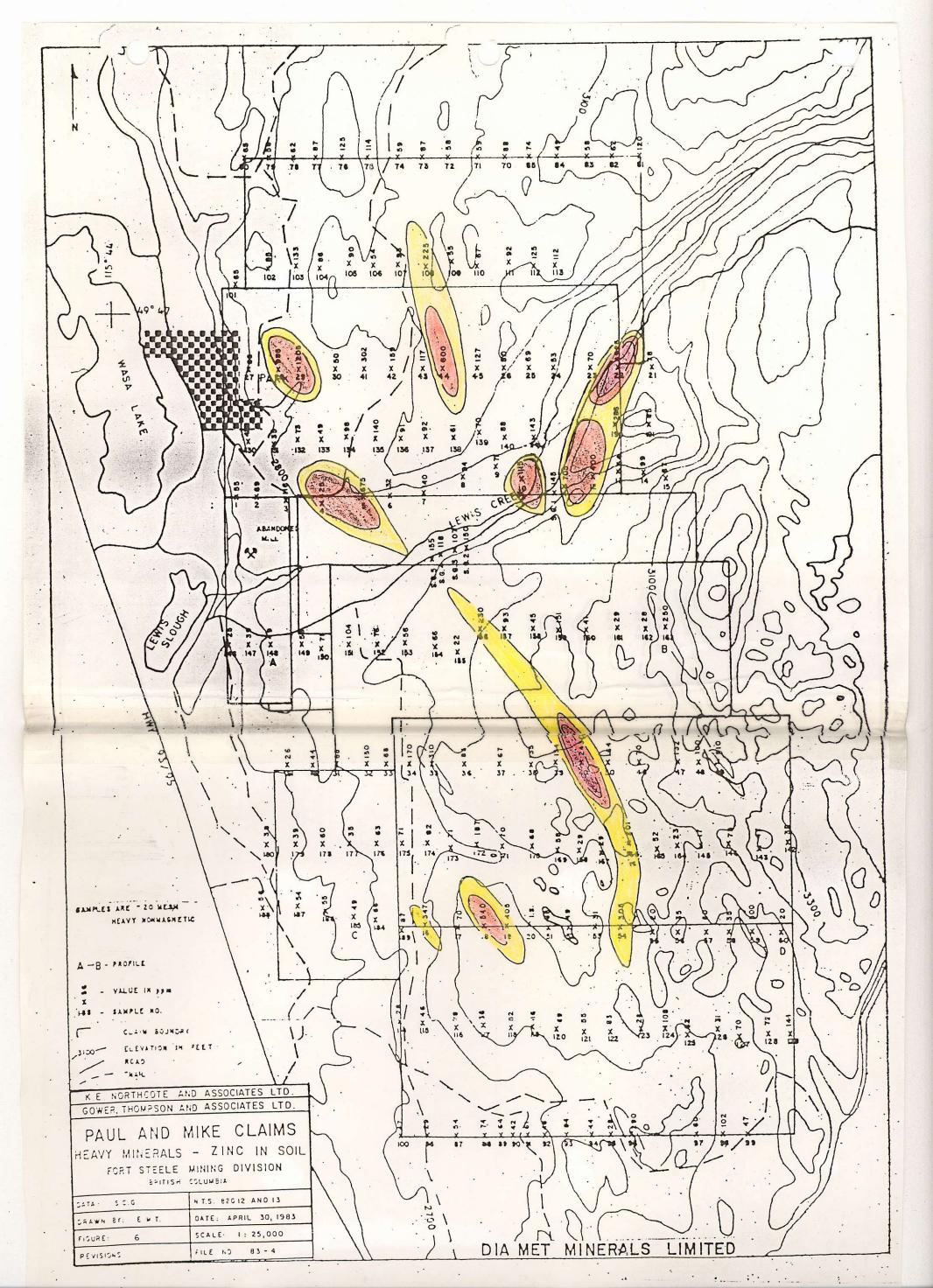
Results

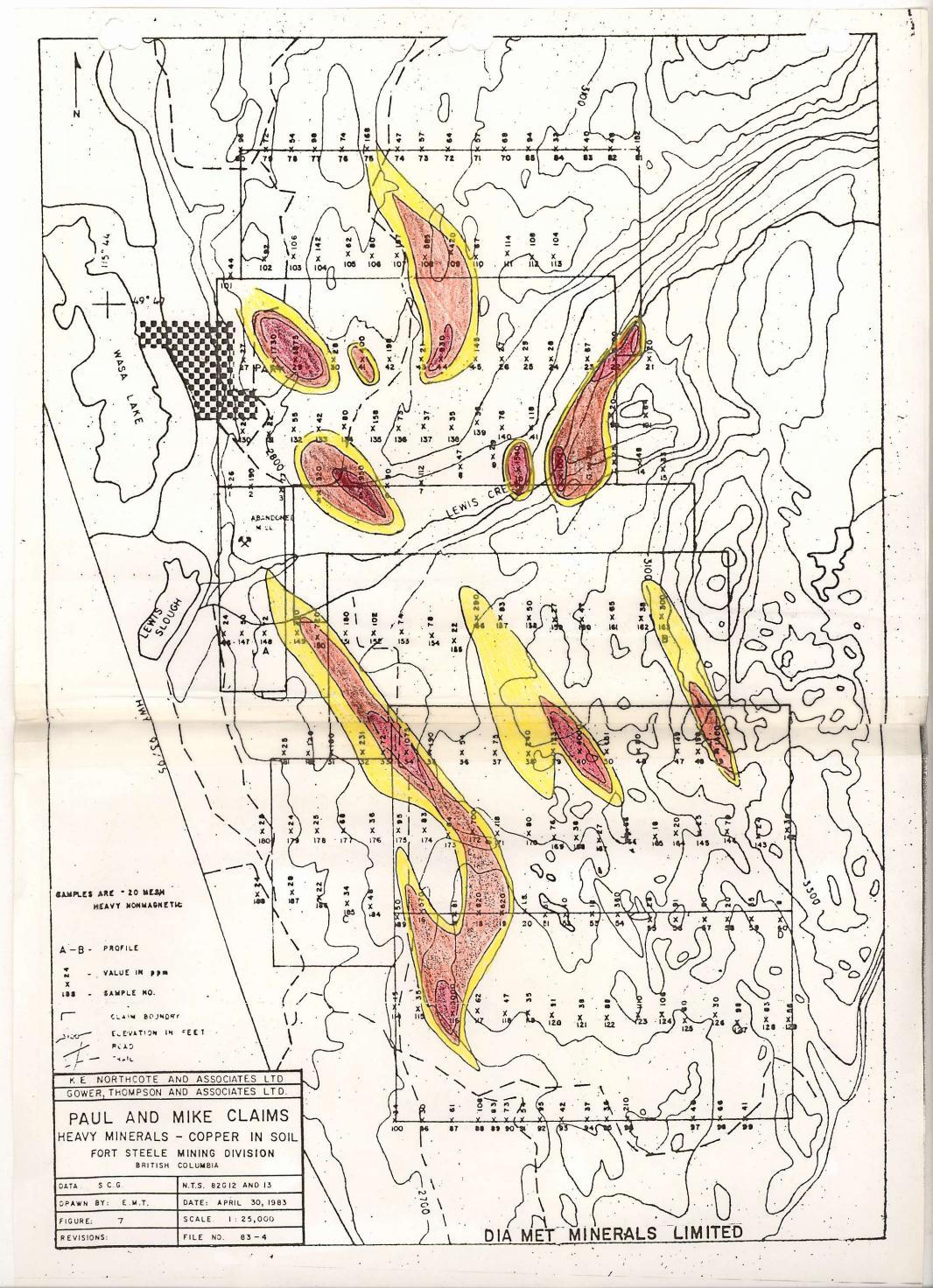
Lead values in the -20HP fraction range from 30 to more than 10,000 ppm with disrupted northwesterly trending highs roughly paralleling resistivity and magnetic trends (Figures 4 and 5). Two profiles were prepared relating lead values to topography, resistivity and magnetics (Figure 9).

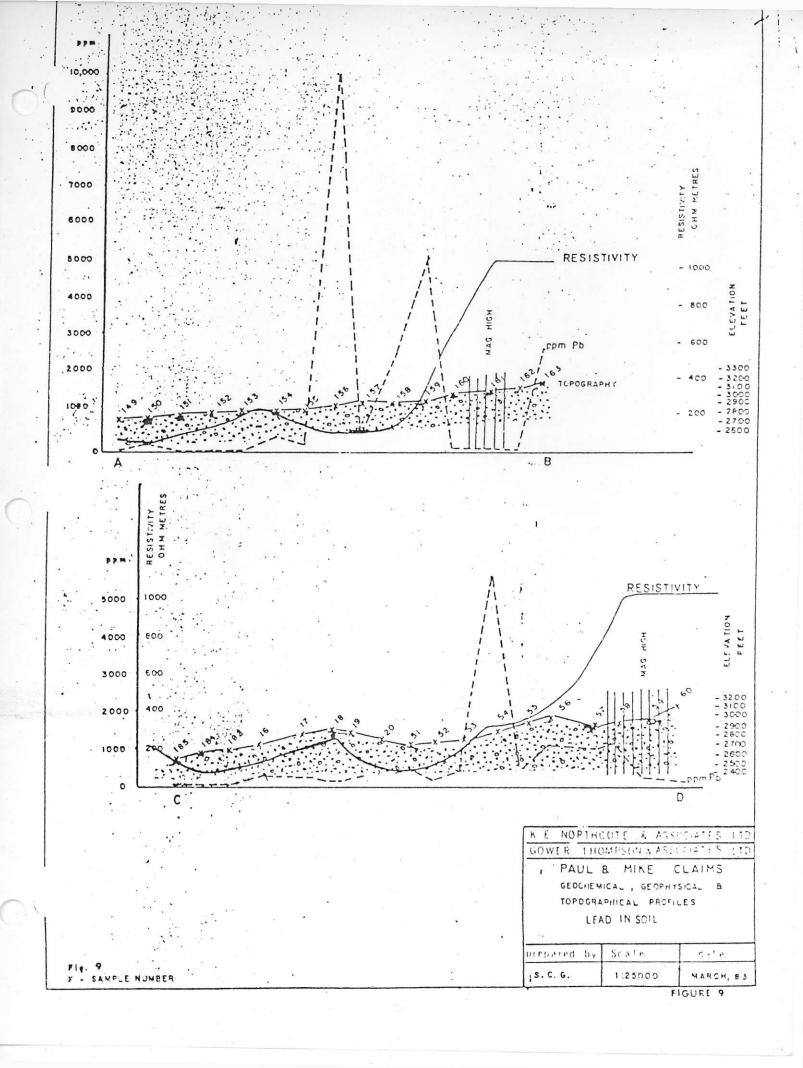
Zinc values in the -20HN fraction range from 17 to 4925 ppm with higher values tending to occue as isolated or clusters of two or three samples. The strongest zinc occurs in the vicinity











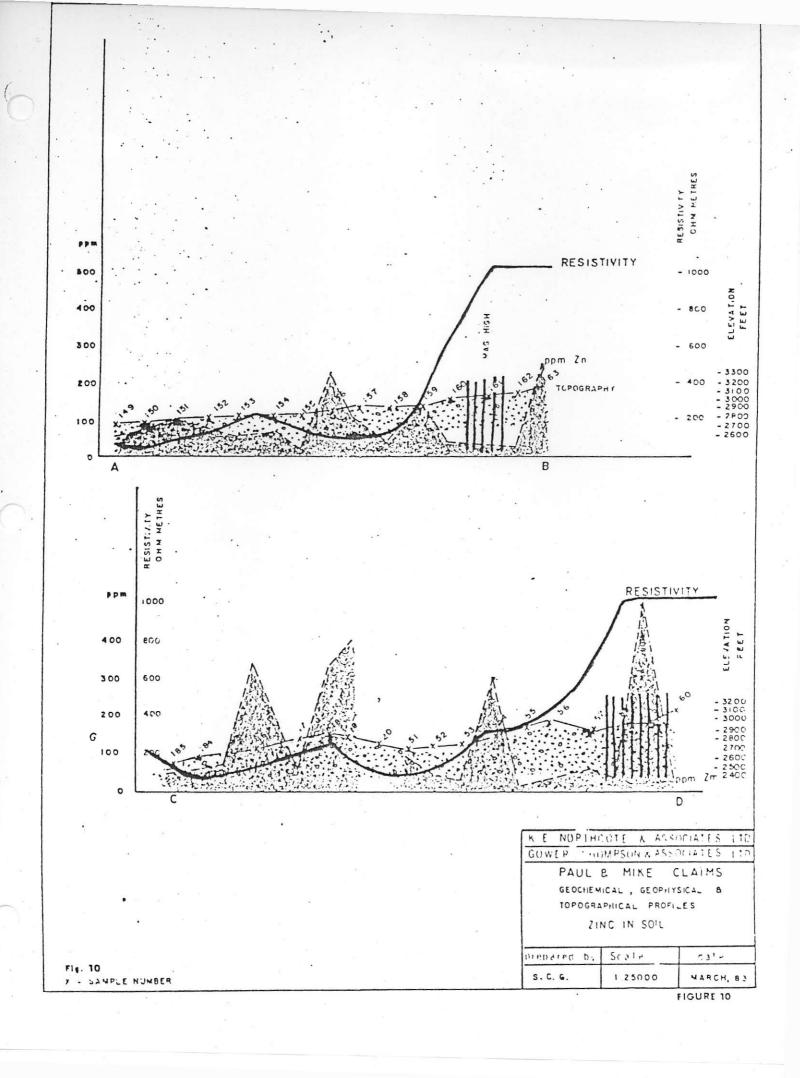
of the Lewis Creek road near outcrops of Fort Steele formation. Because these samples were collected from depths of one to two feet and at some distance from the road, contamination by spillage from ore trucks seems unlikely. Sample SG-1 collected by Gower and Northcote, gives values of 145 ppm Zn which separates two areas of anomalous Zn (Figure 6). Two profiles were prepared relating zinc values to topography, resistivity and magnetics (Figure 10).

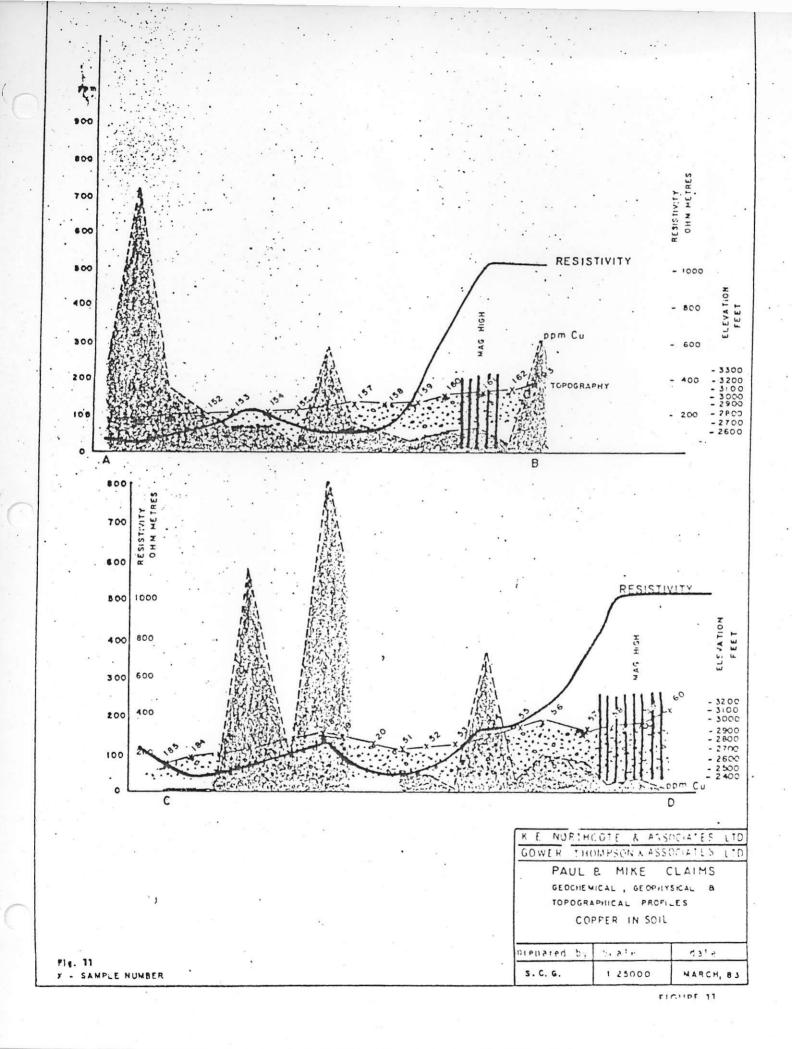
Copper values in the -20HN fraction range from 8 to 8700 ppm occurring as isolated single or double sample highs showing some tendency for northwesterly elongate trends. Two profiles were prepared relating copper values to topography, resistivity and magnetics (Figure 11).

Values for other elements have not been plotted. The samples collected by Gower and Northcote were analyzed for gold and barium at Fipke's request. The background for gold in heavy media samples from this general area is 20 to 50 ppb. Values ranging from 445 to 5855 ppb, with insufficient sample for analysis of S.G.3, indicates anomalous gold concentrations. Barium values are low, ranging from 0.06 to 0.3 percent. High barium values could interfere with values obtained for lead in heavy media samples. (Fipke, personal communication).

Significance of Metal Values in Heavy Media Soil Samples

Discussion between C. Fipke and Northcote centres around significance of metal values in heavy media surface soil samples from areas of overburden of unknown depth and stratigraphy. Fipke suggests that anomalous metal values of heavy media





samples on the Paul and Mike claims could well be reflecting sulphide mineralization from directly beneath. In particular he cites the coincident geophysical and geochemical anomalies in the southeast part of the claim block. Northcote agrees that this is a first priority target area but the suggestion that heavy media metal values from surface soils is reflecting bedrock mineralization might be more acceptable if depth of overburden and its stratigraphy were known. It is difficult to perceive a situation whereby bedrock mineralization can be reflected in surface soil samples through an intervening maximum of 100 metres of interbedded glacial materials including possible intervening protective tills above a basal till overlying and protecting bedrock. Even in basal tills the usual exploration procedure is to follow the metals tail in a up-ice This discussion merely points out the necessity of determining thickness of overburden and its nature prior to final interpretation of metal values from existing surveys and prior to conducting fill-in surveys in present indicated anomalies.

DISCUSSION OF DRILL HOLE GEOCHEMICAL RESULTS

Appendixed are geochemical results from heavy metal concentrates obtained from samples collected from Bearcat Hole 3 for the intervals 600 - 780 feet* and assayed in 10 foot intervals. The Bondar Clegg samples were obtained from the non magnetic fraction whereas the Barringer samples were obtained from the para-magnetic fraction. Thirteen out of seventeen of the magnetic fraction equalled or exceeded 10,000 ppm (>1.0%) and when assayed gave values ranging from 0.43 to 3.66% lead. The paramagentic fraction ranged from 250 to 29,165 ppm (0.025 to 2.91% lead). Silver values ranged from less than 0.2 ppm to 21 ppm in the non-magnetic fraction to less than 0.2 ppm to 145 ppm in the para-magnetic fraction. Zinc values were relatively low. Microscopic examination of the heavy metal concentrates showed that the high lead content was due to a peculiar pellet shaped lead particulates. Studies are continuing to ascertain the origin of what appears to be an organically derived lead rich mineral.

It can be concluded however that anomalous lead and silver are present over a substantial interval in a drill hole. The coincidence of a geophysical anomaly with the geochemical anomaly is noteworthy and warrants further study.

^{*} This hole did not reach bedrock. It was cased and can be re-entered and deepened.

RESULTS OF 1988 ROTARY MUD DRILLING

In December of 1987 two holes were collared with a rotary air drill, drilled and cased to 270 feet, well into the glacial clays which proved so difficult to drill through in the 1885 program.

A rotary mud drill was then employed to drill through the overburden sediments and provide a cased hole which would allow diamond drilling of bedrock. The rotary mud drill was effective in getting to bedrock at about 915 feet in the first hole. The hole was continued to 938 feet in bedrock.

Collection of drill cuttings was very inefficient with the mud drill as the clay and silt encountered in the hole readily went into suspension. In the shallower parts of the holes small chips of clay were recovered but for much of the drilling only a very small sample of clay, silt and occasional coarse sand was collected. Toward the bottom of the second hole, a number of larger boulders were encountered which produced noticeable jarring of the drill rig.

The second hole was drilled to 1991 feet; productivity was particularly slow in the lower 50 feet or so and bedrock may have been encountered within this interval, but few rock chips were recovered on surface. Technical problems encountered while attempting to lower the casing to bottom after the drilling was completed confirmed earlier suspicions that the hole walls were washing out. This washing out of the hole walls may have resulted in sufficient decrease in the turbulence of the return drilling fluid to effectively prevent most "bedrock" chips from being returned to surface.

The rock chips collected from Hole 1 are predominantly of two lithologies. Fine to medium-grained white quartzites are probably from the Fort Steele Formation. Finer-grained dark blue-gray siltstones are similar to lithologies seen in the Middle Aldridge Formation.

In Hole 2 there are relatively few angular chips that are similar in physical character to the bedrock chips collected from Hole 1. These are mostly of fine-grained, dark blue-gray siltstone which is a common lithology of the Middle Aldridge Formation.

The drill results are thus compatible with an eastern fault zone that separates Fort Steele Formation on the east from Middle Aldridge Formation on the west.

If the second, western rotary mud drill hole did get to bedrock and the interpretation of Middle Aldridge Formation is correct, then Lower-Middle Aldridge stratigraphy exists below the Paul - Mike claim group and the anomalous lead-silver mineralization seen in overburden on the property may be from a bedrock source located at a stratigraphic position comparable to that of the Sullivan orebody.

The deepest hole drilled in 1985, to 780 feet, provided significant anomalous lead and silver results. This hole was deepened in 1988 but technical problems encountered early in the drilling prevented reaching bedrock; the hole terminated at 960 feet. Samples were collected at 20 foot intervals during the drilling but very little sample material was collected. Evidently the clay and silt material being drilled broke up rapidly and went into suspension, preventing a simple gravity separation of the drill cuttings from the return flow of drilling mud.

Geochemical analyses of the heavy mineral fraction from this drill hole (W-88-3 in the appendix) show a rapid decrease in the copper, lead, zinc and silver values (with the exception of a high silver at the hole bottom). These results are in strong contrast to the results from the upper part of the hole drilled with a rotary air rig in 1985; between 600 and 780 feet most sample intervals had more than 10,000 ppm lead in the heavy mineral fractions. This variation in results between the 1985 and 1988 drilling is considered to be a function of the drilling method. The 1985 drilling was with an air riq while the 1988 drilling was with a mud rig. The spherical lead particles present in the overburden clays are probably not being returned to surface during drilling with the mud drill. It is evident from the nature of the return circulation that the clay and silt was being completely broken up in the drill mud and most of the spherical-shaped lead particles would preferentially stay near the bottom of the hole and not be returned to surface.

In summary, anomalous lead and silver geochem results from the work up to 1985, and favourable geological results from the 1988 drilling infer a possible lead-zinc-silver deposit below glacial overburden on the Paul - Mike claims.

PROPOSED EXPLORATION PROGRAM

Geochemical results obtained from near-surface overburden drilling and limited geological results interpreted from chip samples of bedrock in one drill hole and possible bedrock in a second drill hole support the existence of a buried lead-zinc-silver deposit on the Paul - Mike claims.

Further work is warranted to evaluate the anomalous geochemistry, substantiate the bedrock geology and test for a buried sulfide deposit.

It is recommended that additional drilling be undertaken to test both the overburden material and the bedrock below the anomalous geochemical results. An additional two holes should be drilled from surface to bedrock. Bedrock should then be tested in these holes and the available previously drilled holes to test for a buried sulfide deposit. Estimated cost of the drill program is \$250,000.

ESTIMATED BUDGET OF PROGRAM

| Drilling Rotary mud drilling surface to bedrock | |
|--|--------------|
| 4000 feet @ \$25.00/foot | \$100,000.00 |
| Diamond drilling bedrock | |
| 1200 feet @ \$50.00/foot | 60,000.00 |
| Mobilization and demobilization of drill rigs | 20,000.00 |
| Geochemistry | |
| Sample preparation and analyses of heavy minerals 150 samples @ \$100.00/sample | 15,000.00 |
| Geochemical analyses and assays of drill core 100 samples @ \$30.00/sample | 3,000.00 |
| Geological Supervision 40 days @\$350.00/day | 14,000.00 |
| | |
| Report and Consulting | 3,000.00 |
| Technical support | |
| 40 days @ \$200.00/day | 8,000.00 |
| Accomodation and vehicle rental | |
| 80 man days @ \$150.00/day | 12,000.00 |
| | \$235,000.00 |
| 6% Contingency | 15,000.00 |
| TOTAL BUDGET | \$250,000.00 |

APPENDIX

- A. Bondar Clegg Geochemical Lab Report with analyses of Bearcat Explorations drill hole W-85-3 for the interval 600 780 feet.
- B. Barringer Magenta Geochemical Report with analyses of Bearcat Explorations drill hole W-85-3 for the interval 600 780 feet.
- C. Barringer Magenta Geochemical Lab Report for analyses of 1988 drill holes.

REPORT: 127-1026 (COMPLETE:)

REFERENCE INFO:

CLIENT: DIA MET MINERALS LTD.
PROJECT: NONE GIVEN

SUBMITTED BY: CF MINERALS DATE PRINTED: 16-APR-87

NUMBER OF LOWER ORDER -ELEMENT ANALYSES DETECTION LIMIT EXTRACTION METHOD. 5 PPB 11 Au Gold NOT APPLICABLE IND. NEUTRON ACTIV. 17 D.2 PPH NOT APPLICABLE IND. NEUTRON ACTIV. 2 Sb Antimony 17 1 PPM NOT APPLICABLE IND. NEUTRON ACTIV. As Arsenic Ba NOT APPLICABLE IND. NEUTRON ACTIV. Barium 17 100 PPM 17 10 PPM NOT APPLICABLE IND. NEUTRON ACTIV. Cá Cadmium 17 NOT APPLICABLE IND. NEUTRON ACTIV. Cs Cesium 1 PPM 17 50 PPM NOT APPLICABLE . 7 Cr Chromium IND. NEUTRON ACTIV. 8 . Co . NOT APPLICABLE Cobalt 117 10 PPM IND. NEUTRON ACTIV. 9 Eu Europium 17 2 PPH NOT APPLICABLE IND. NEUTRON ACTIV. IND. NEUTRON ACTIV. NOT APPLICABLE 10 Hf. Hafnium! 17 2 PPM Ir Iridium 17 100 PPB NOT APPLICABLE IND. NEUTRON ACTIV. 12 Fe? Iron 17 D.5 PCT NOT APPLICABLE IND. NEUTRON ACTIV. 17 5 PPM NOT APPLICABLE IND. NEUTRON ACTIV. 13 Lanthanum La: IND. NEUTRON ACTIV. 17 2 PPM NOT APPLICABLE 14 No Molybdenum NOT APPLICABLE IND. NEUTRON ACTIV. 15 Ni Nickel 17 50 PPH 17 IND. NEUTRON ACTIV. Rb Rubidium' 10 PPM NOT APPLICABLE 16 IND. NEUTRON ACTIV. NOT APPLICABLE 17 Sc Scandium -17 0.5 PPM NOT APPLICABLE IND. NEUTRON ACTIV. 18 . Se Selenium 17 10 PPM 17: 5 PPH NOT APPLICABLE IND. NEUTRON ACTIV. 19 Ag Silver IND. NEUTRON ACTIV. 20 Ta Tantalum 17. 1 PPM NOT APPLICABLE 1 PPM NOT APPLICABLE IND. NEUTRON ACTIV. Tb 17. 21 Terbium IND. NEUTRON ACTIV. NOT APPLICABLE 22. Th Thorium D.5 PPM IND. NEUTRON ACTIV. 2 PPM NOT APPLICABLE 117 23 Tungsten IND. NEUTRON ACTIV. U :17 0.5 PPM NOT APPLICABLE 24 Uranium IND. NEUTRON ACTIV. 17 5 PPM NOT APPLICABLE Yb 25 Ytterbium IND. NEUTRON ACTIV. .17 200 : PPM NOT APPLICABLE 26 Zn Zinc . HN03-HCL HOT EXTR Atomic Absorption 17 1 PPH 27 Cu Copper HN03-HCL HOT EXTR Atomic Absorption 17 2 PPM Pb 28 Lead 17 1 PPM HN03-HCL HOT EXTR Atomic Absorption Zinc 29 · Zn HN03-HCL HOT EXTR Atomic Absorption 17 D.1 PPM 30 Silver

ar-Orgs & Company Ltd. Geochemical North Vancouver, B.C. Canada V7P 2R5 Lab Report Phone: (604) 985-0681 Telex: 04-352667 REPORT: 127-1026 (COMPLETE) REFERENCE INFO: CLIENT: DIA MET MINERALS LTD.
PROJECT: NONE GIVEN SUBMITTED BY: CF MINERALS DATE PRINTED: 16-APR-87 SAMPLE TYPES NUMBER SIZE FRACTIONS. NUMBER SAMPLE PREPARATIONS . NUMBER : PULVERIZING C CONCENTRATE (PAN/HM) 17 17 BATCH SURCHARGE NOTES: 6 indicates SMALL SAMPLE WEIGHT REMARKS: PLEASE CONTACT US IF ASSAYS ARE NEEDED ON INVOICE TO: DIAMET MINERALS REPORT COPIES TO: DIAMET MINERALS

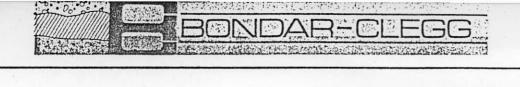
130 Pemberton Ave. North Vancouver, B.C. Canada V7P 2R5 Phone: (604) 985-0681 Telex: 04-352667



Geochemical Lab Report

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|---|-------------------|------------------------------|--|------------------|----------------------------|--------------------|------------------|----------------------|-------------------|-----------------|-----------------|----------------------|
| SAMPLE NUMBER | ELEMENT. UNITS | Au S PPB PP | A CONTRACTOR OF THE PARTY OF TH | As PPH | Ba PP# | Cd PPII | Cs PPM | Cr PPM | Co PPM | Eu PPM | Hf PPM | Ir PPB |
| C1_0B 20-60 C1_0B 21-61 C1_0B 22-62 | 0'-620 | <140 17. 540 10. 64 61 | o to | 180 42 134 | /19000 /-3200 /-7900 | <190 <42 <39 | <21 <4 | <770 <100 | <50 <20 | <42 <10 | 430 100 | <1300 |
| C1 0B 22-62 C1 0B 24-64 | 0'-640 | . <11 | Ż . | 31. 51. | 2800 2600 | <10 <51 | <4 2 2 4 4 | 140 71 <100 | 93 31 <20 | 15 5 <9 | 336 72 80 | <240 <100 <200 |
| C1 0B 25-65 | 0'-670 | . <25 | 0.700 | 21 | ∵790 № 2100 | <50 <50 | <5 <5 | <250 <250 | <50 <50 | | 17 | <500 <500 |
| C1 0B 27-67 C1 0B 28-68 C1 0B 29-69 | 0'-690 | <57 18. 59 2. 30 2. | 9 | 18 13 | <500 7400 7900 | <100 <27 <20 | <5 <2 6 | <250 <100 <100 | <50 <20 <20 | <10 <4 <4 | 30 20 | <500 <200 <200 |
| C1 0B 30-70 C1 0B 32-72 | | 1200 1.4 | | 16 14 | 2400 1100 | <1D <10 | <1 3 | <50 93 | 16 | <2 <2 | 20 12 | <100 <100 |
| C1 0B 33-73 C1 0B 34-74 C1 0B 35-75 | D'-750 | | | 8 14 120 | 870 1000 25000 | <10 <10 <260 | 4 5 <22 | 53 97 <500 | 13 <10 <100 | <2 4 <72 | 8 9 370 | <100 <1000 |
| C1 08 36-76 C1 08 37-77 | | 1170 5. <25 3.6 | | 67 53 | 11000 >30000 | <42 <300 | 4 <5 | 160 <250 | 56 .<50 | <9 <10 | 200 140 | <200 <500 |
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Geochemical Lab Report

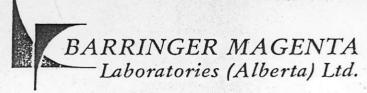
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|-------|--------|---------------------|---------------------------------------|-----------|-------------|-----------|------|--------|-----------|--------------|-----------|-----------|---------------|--------|
| | SAMPLE | | ELEMENT UNITS | Fe PCT | La PPM | Mo PPM | | Rb PPM | Sc PPM | Se PPM | Ag PPM | Ta PPM | Th PPM | Th PPN |
| 11 | C1 0B | 20-600'-61 | 0 | 15.0 | 72D | <51 | <480 | <250 | 33.0 | <220 | <120 | 45 | 21 | 170.0 |
| 4-1-4 | C1 0B | 21-610'-62 | 0 | 3.2 | 210 | 8 | <130 | <62 | 16.0 | <20 | 27 | 7 | 5 | 53.0 |
| | C1 0B | 22-620'-63 | 0 | 11.0 | 720 | <11 | <110 | <57 | 37.0 | <42 | 48 | 31 | 13 | 147.0 |
| | C1:08 | 23-630'-64 | 0 | 3.7 | 190 | <2 | <50 | 41 | 13.0 | <10 | <5 | 6 | 4 | 38.0 |
| | C1 0B | 24-640'-65 | 0 | 7.2 | 210 | <11 | <130 | <74 | 13.0 | <20 | 53 | 8 | 5 | 45.0 |
| | C1 0B | 25-650'-66 | 0 | <2.5 | 30.8 | <10 | <100 | <50 | 6.0 | <50 | 26 | <5 | < 5 | 7.0 |
| | C1 0B | 26-660'-67 | 0 | <2.5 | 53 | <10 | <100 | <50 | 4.7 | <50 | 47 | . <5 | <5 | 15.0 |
| | C1 0B | 27-670'-68 | 0 | <2.5 | . 88 | <25 | <100 | <110 | 6.3 | <50 | 170 | <,5 | <5 | 23.0 |
| | C1 0B | 28-680'-69 | 0 | 2.9 | 75 | <4 | <50 | <20 | 7.6 | <20 | 26 | <2 | 3 | 23.0 |
| | C1 OB | 29-690'-70 | 0 | 2.5 | 55 | <4 | <50 | <20 | 9.2 | <20 | <10 | <2 | <2 | 17.0 |
| | C1 0B | 30-700'-71 | 0 | 2.5 | 79 | <4 | <50 | 43 | 7.2 | <10 | <5 | <1 | 2 | 18.0 |
| | C1 08 | 32-720'-73 | 0 | 3.4 | 57 | <2 | <50 | 70 | 8.3 | <10 | <5 | <1 | <1 | 15.0 |
| | C1 0B | 33-730'-74 | 0 | 2.7 | 45 | <2 | <50 | 43 | 7.6 | <10 | 7 | <1 | <1 | 13.0 |
| | C1 08 | 34-740'-75 | 0 | 2.7 | 48 | <2 | <50 | 64 | 8.1 | <10 | <5 | <1 | <1 | 14.0 |
| · | C1 0B | 35-750'-76 |) | 11.0 | 600 | <74 | <480 | <310 | 20.0 | <240 | 280 | 33 | 28 | 130.0 |
| | C1 0B | 36-760 '- 77 | 0 | 4.9 | 380 | <11 | 150 | <61 | 16.0 | <20 | 22 | 16 | -8 | 86.0 |
| | C1 0B | 37-770'-78 |) | <2.5 | 200 | <10 | <100 | <50 | 11.0 | <50 | . 33 | 10 | <5 | 49.0 |



Geochemical Lab Report

PAGE 10

| REPORT: 127- | -1026 | | | | | | | PROJECT: | NONE GIVEN |
|------------------|---------|-----|--------------|-------|-------|------|--------|----------|------------|
| SAMPLE | ELEMENT | H | Problem U.S. | Yb | Zn | Cu | РЬ | Zn | Ag |
| NUMBER | UNITS | PPM | PPH | PPM | PPM | PPH | PPH | PPM | HAA |
| C1 0B 20-600 |)'-61D | 180 | 72.0 | <25 | <1000 | 259 | >10000 | 1210 | . 10.0 |
| C1 08 21-610 | 7'-620 | 24 | 18.0 | <10 | 1200 | 322 | >10000 | 985 | 9.2 |
| C1 .0B 22-621 | 0'-630 | 39 | 46.0 | 43 | <410 | 132 | >10000 | 203 | . 2.1 |
| C1 0B 23-630 | 0'-640 | <5 | 12.0 | 9 | <200 | 44 | 4000 | 79 | <0.2 |
| C1 0B 24-640 | 0620 | 14 | 21.0 | 11 | <980 | 274 | >10000 | 1005 | 20.0 |
| C1 08 25-650 | 1'-660 | <10 | <2.5 | <25 | <1000 | 28 | 1815 | 135 | <0.2 |
| C1 08 26-668 | 0*-670 | <10 | 36.D | <25 | 3500 | 131 | >10000 | 5600 | 6.5 |
| C1 0B 27-670 | o'-68D | 73 | 55.0 | <25 | 4000 | 171 | >10000 | 3530 | 21.0 |
| C1 0B 28-680 | 0'-690 | 27 | 11.0 | <10 | <400 | 84 | >10000 | 500 | 13.0 |
| C1 0B 29-690 | 700 | <4 | 8.5 | <10 | <400 | 83 | >10000 | 353 | 0.7 |
| C1 0B 30-700 |)'-710 | 13 | 5.2 | 6 | <200 | . 54 | 7400 | 92 | 0.4 |
| C1 08 32-720 | '-730 | 4 | 5.5 | <5 | 520 | 76 | >10000 | 465 | 0.2 |
| C1 0B 33-730 |)*-74U | <2 | 3.6 | <5 | <200 | 41 | 3400 | 73 | <0.2 |
| C1 OB 34-740 | 7-750 | 5 | 4.4 | <5 | <200 | 43 | 10000 | 127 | <0.2 |
| C1 08 35-750 | 1'-760 | 140 | 75.0 | <50 . | <2000 | 260. | >10000 | 2250 | <2.00 |
| C1 OB 36-760 | -770 | 15 | 30.0 | 12 | <400 | 143 | >10000 | 560 | 0.8 |
| C1 OR 37-770 | '-780 | 89 | 28.0 | <25 | <1000 | 351 | >10000 | 298 | n.3 |



AUTHORITY: C. FIPKE

4200B - 10 STREET N E. CALGARY, ALBERTA T2E 6K3

23. PHONE. (403) 250-1901

TAGE: 1 00 1 COPY: 1 07 2

DIA MET MINERALS LTD. 1675 FOWICK RD. KELOWNA, B.C. VIX 4L1

WORK ORDER: 4060D 82

AAA FINAL REPORT AAA

GEOCHEMICAL LABORATORY REPORT

| S | AMPLE | TYI. | C: | 1 12 | JY | MINERAL SET | | |
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| 03 | 21:310 | .520 | | | | 2730.0 | <5.0 | 30.0 |
| OE | 22:620 | 0.000 | | | | 1800.0 | <5.0 | 2.1 |
| 0.3 | 23:630 | -540 | | | | 330.0 | <5.0 | < 0 . 2 |
| OB | 24:640 | . 650 | | | | 5500.0 | <5.0 | 24.0 |
| | | | | | | | | |
| 1 | 25:650 | | | | | 250.0 | <5.0 | 3.0 |
| | 20:000 | | | | | 14655.0 | <5.0 | <0.2 |
| 1 | 27:670 | | | | | 29165.0 | <5.0 | <0.2 |
| OB | 20:600 | . 620 | | | | 3160.0 | <5.0 | < 0.2 |
| 03 | 20:600 | -700 | | | | 7500.0 | <5.0 | <0.2 |
| nr. | 30:700 | . 710 | | | | 3125.0 | <5.0 | 2.1 |
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| OD: | 35:750 | . 760 | | | | 10500.0 | <5.0 | <0.2 |
| 1 11:4 | 36:760 | .770 | | | | 2300.0 | <5.0 | 27.0 |
| 2000 2000 | 37:770 | | | | | 3090.0 | <5.0 | 13.8 |
| 0.15 | W/ W// V | 100 | | | | O V J V R V | 7 W # W | at. 1 1 |

SIGNED:

C. Bouglas Read,

LABORATORY MANAGER

Para-magnetic

FOOTHOTES:

PROJECTIONABLE PRECISION: ARINTERPERUNCE: TRATENCE: NEWS DESECTION: 18-1009771CIENT SAMPLE; NARNOT ANALYZED: NSEGNOSING SAMPLE:

| HOLE | DEPT H | ANA | | 0 HN RESULTS | (PPM) | | | ANALYTIC | -60 HP AL RESUI | TS (PPM) | |
|-----------------|---------------|------|-----|-----------------|-------|-----|------|----------|--------------------|----------|-----|
| | (Feet) | Cu | Pb | Zn | Ag | Mo | . Cu | Pb | Zn | Ag | Мо |
| DH-1-87 | 820-900 | 94 | 85 | 113 | 0.2 | <1 | 93 | 710 | 255 | 40.2 | 18 |
| | 900-910 | 120 | 110 | 141 | 0.4 | <1 | 53 | 205 | 151 | < 0.2 | < 1 |
| | 910-920 | 48 | 80 | 254 | 0.3 | <1 | 69 | 184 | 127 | < 0.2 | 32 |
| | 920-930 | 276 | 245 | 168 | 2.5 | 4 | 95 | 217 | 139 | <0.2 | 81 |
| 5. ° | 930-940 | 625 | 535 | 117 | 1.9 | - 5 | 94 | 283 | 119 | <0.2 | 119 |
| DH-2-87 | 1880-1900 | 220 | 650 | 564 | 0.6 | 41 | 40 | 348 | 178 | 0.5 | <1 |
| | 1900-1920 | 247 | 363 | 297 | 0.5 | 12 | 37 | 302 | 140 | 3.6 | <1 |
| | 1920-1940 | 203 | 560 | 603 | 0.3 | 2 | 43 | 535 | 227 | 0.7 | |
| | 1940-1960 | 154 | 490 | 508 | 0.9 | 2 | . 93 | 453 | 182 | 0.4 | 2 |
| | 1960-1986 | 1403 | 358 | 109 | 2.3 | 3 | 325 | 280 | 250 | 0.2 | 4 |
| | 1980-1990 | 255 | 367 | 254 | 0.7 | <1 | 92 | 268 | 21€ | 1.5 | <1 |
| DH-3-87 | 350-360 | | | | | | 117 | 671 | 635 | 0.2 | 42 |
| | 360-370 | | | | | | 118 | 809 | 358 | 5.2 | 71 |
| | 370-380 | | | | | | 261 | 1037 | 306 | 4.5 | 249 |
| | 380-390 | Ì | | | | | 1720 | 880 | 1432 | 3.4 | 54 |
| | 390-400 | | | | | | 322 | 324 | 419 | 10.3 | 105 |
| | | 5. | | | | | | | | | |

| HCLE | DEPTH (Ft.) | | | H ASSAY R | N FSIII.TS | | | | λςςλ | HP Y RESULT | ne | | |
|--------|--|---|---|---|---|---|---|----------------------------|---|---------------------------------|---|---|---|
| | (1 0.) | Cu PPM | Pb PPM | Zn PPM | Ag PPM | Mo PPM | Au PPB | Cu PPM | Pb PPM | Zn PPM | Ag PPM | Mo PPM | |
| W-85-3 | 600 - 610' 610 - 620' 620 - 630' 630 - 640' 640 - 650' 650 - 660' 660 - 670' 670 - 680' 680 - 690' 700 - 710' 720 - 730' 730 - 740' 740 - 750' 750 - 760' 770 - 780' | 171 84 83 54 76 41 43 260 143 | >10,000 36,600 29;800 4,300 6,700 1,815 >10,000 >10,000 28,000 28,000 28,000 26,900 3,300 9,600 >10,000 >10,000 >10,000 | 1210 985 203 79 1005 135 5600 3530 500 353 92 465 73 127 2250 560 298 | <pre><120 27 48 < 5 53 26 47 170 26 < 10 < 5 < 5 280 22 33</pre> | <pre><51 8 <11 < 2 <11 <10 <10 <25 < 4 < 4 < 2 < 2 < 74 <11 <10</pre> | <pre>< 140 540 64 < 11 52 < 25 < 25 < 57 59 30 1200 12 22 5 240 1170 < 25</pre> | | 3500 2730 1800 880 5500 250 14655 29165 3160 7500 3125 2475 1580 5110 10500 2800 3090 | | 145.0 30.0 2.1 <0.2 24.0 3.0 <0.2 <0.2 <0.2 <0.2 <0.2 2.1 3.3 0.8 3.6 <0.2 27.0 13.8 | <pre>\$5 \$5 \$5 \$5 \$5 \$5 \$5 \$5 \$5 \$5 \$5 \$5 \$5 \$</pre> | • |
| ₩-88-3 | 860 - 880' 880 - 900' 900 - 920' 920 - 940' 940 - 960' | 349 273 298 197 228 | 3,975 1,925 1,102 350 752 | 1023 697 1423 357 306 | 2.1 0.9 0.6 0.3 4.7 | <1 <1 <1 <1 5 | | 90 79 88 81 57 | 1425 647 374 238 221 | 745 508 346 206 177 | 0.3 0.2 40.2 40.2 0.4 | 6 13 9 8 < 1 | |

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AUTHORITY: C. FIPKE

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*** FINAL REPORT ***

| | GE | OCHI | EMICAL | LABO | DRATORY | REPOR | x · |
|--------------|---|-------------------------|--------------------------------------|---|--|---|-------------------------------------|
| | MPLE H P L | | CONCENTR M B E R | ATE CU PPM | PB PPM | ZN PPM | MO PPM |
| M-88 M-88 | -3:86 -3:88 -3:90 -3:92 | 0-900 0-920 0-940 | 20HN 20HN 20HN 20HN 20HN | 349.0 273.0 298.0 197.0 228.0 | 3975.0 1925.0 1102.0 350.0 752.0 | 1023.0 697.0 1423.0 357.0 306.0 | <1.0 <1.0 <1.0 <1.0 5.0 |
| W-88 W-88 | -3:86 -3:88 -3:90 -3:92 -3:94 | 0-900 0-920 0-940 | 20HP 20HP 20HP 20HP 20HP | 90.0 79.0 88.0 81.0 57.0 | 1425.0 647.0 374.0 238.0 221.0 | 745.0 508.0 346.0 206.0 177.0 | 6.0 13.0 9.0 8.0 <1.0 |

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WORK ORDER: 51180-88

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GEOCHEMICAL LABORATORY REPORT

| í | | S | AM | PI | F | T | YP: | E: | • | co | NC | EN' | TRATE |
|---|-----|-----|---------|-----|-----|--------------|-----|----|---|-----|-----|-----|-------|
| | | _ | • • • • | | | _ | | | • | | | | AG |
| | S | Α | М | P | L | E | N | U | H | B | E | R | PPM |
| | Ш | -88 | 3-3 | : 8 | 361 | o-8 | 80 | | | | -20 | инс | 2.1 |
| | - | | _ | | | 0-9 | | | | | | ИНС | 0.9 |
| | W- | -88 | 3-3 | | 00 | 0-9 | 20 | | | | -20 | ИНС | 0.6 |
| | W - | 88 | 3-3 | : 9 | 20 | 0 - 9 | 40 | | | | -2(| ИНС | 0.3 |
| | W | 88 | 3 - 3 | : 9 | 4 | 0-9 | 60 | | | *** | -20 | ИНС | 4.7 |
| | 닖 - | -81 | 3-3 | : 8 | 36 | 0-8 | 80 | | | | -20 | HP | 0.3 |
| | | | - | | - | 0-9 | | | | | -2(| HP | 0.2 |
| | | | | | | 0-9 | | | | | -2(| ?H₽ | <0.2 |
| | W- | -88 | 3-3 | : 9 | 20 | 9-0 | 40 | | | | -20 | HP | <0.2 |
| | W- | -88 | 3-3 | : 5 | 4 (| 0-9 | 60 | | | | -2(| 4HC | 0.4 |
| | | | | | | | | | | | | | |

SIGNED: C. Douglas Read,

LABORATORY MANAGER

PEQUESTIONABLE PRECISION; A=INTERFERENCE; TRETRACE; NUENOT DETECTED; IS=INSUFFICIENT SAMPLE; NA=NOT ANALYZED; MS=MISSING SAMPLE

ADVANCED TECHNIQUES AND INSTRUMENTATION FOR THE EARTH SCIENCES

B.C. VIX 4L1

WORK ORDER: 51180-88

*** FINAL REPORT ***

GEOCHEMICAL LABORATORY REPORT

| SAMPLE TYPE: CONCENTRATE CU PB ZN MO PPM PPM PPM PPM PPM PPM PPM PPM PPM | | | | | | • |
|--|-----------------------|---------|--------|--------|--------|---------------------------------------|
| BH287:1980-1990 | SAMPLE TYPE: | CONCENT | RATE | | | |
| SAMPLENUMBER PPM P | | | CU | PB | ZN | MO |
| DH287:1900-1920 | SAMPLE NU | MBER | PPH | PPM | | |
| DH287:1900-1920 | DH287:1880-1900 | -60HP | 40.0 | 348.0 | 178.0 | <1 ₋ 0 |
| DH287:1920-1940 | | | | | | |
| DH287:1940-1960 | | | | | | |
| DH287:1960-1980 | | | | | | |
| DH287:1880-1900 | DH287:1960-1980 | -GOHP | | | | |
| DH287:1880-1900 | İ | | | | | |
| DH287:1900-1920 | | | | | | |
| BH287:1920-1940 -60HN 203.0 560.0 603.0 2.0 BH287:1940-1960 -60HN 154.0 490.0 508.0 2.0 BH287:1960-1980 -60HN 1403.0 358.0 109.0 3.0 LH287:1980-1990 -60HN 255.0 367.0 254.0 <1.0 | | | | | | |
| DH287:1940-1960 | | | | | - | 12.0 |
| BH287:1960-1980 -60HN 1403.0 358.0 109.0 3.0 IH287:1980-1990 -60HN 255.0 367.0 254.0 <1.0 | | | | - " | | |
| DH287:1980-1990 | DH287:1940-1960 | -GOHN | 154.0 | 490.0 | 508.0 | 2.0 |
| DH287:1980-1990 | #H287:1960-1980 | -60HN | 1407 0 | 358 ለ | 100 () | 3 A |
| DHI-87:890-900 | | | | | | |
| DHI-87:900-910 | | | | | | |
| DHI-87:910-920 -60HN 48.0 80.0 254.0 <1.0 DHI-87:920-930 -60HN 270.0 243.0 168.0 4.0 DHI-87:930-940 -60HN 625.0 535.0 117.0 5.0 DH3-87:350-360 -20HP 117.0 671.0 635.0 42.0 DH3-87:360-370 -20HP 118.0 809.0 358.0 71.0 DH3-87:370-380 -20HP 261.0 1037.0 306.0 249.0 DH3-87:380-390 -20HP 1720.0 880.0 1432.0 54.0 DH3-87:380-390 -20HP 322.0 324.0 419.0 105.0 W-85-2:540-550 -20HP 322.0 324.0 419.0 105.0 W-85-2:550-560 -20HP 121.0 1794.0 55.0 23.0 W-85-2:560-570 -20HP 198.0 288.0 324.0 395.0 W-85-2:580-590 -20HP 231.0 1697.0 415.0 23.0 W-85-2:580-590 -20HP </td <td></td> <td></td> <td></td> <td></td> <td></td> <td></td> | | | | | | |
| BHI-87:920-930 | | | | | | |
| DHI-87:930-940 -60HN 625.0 535.0 117.0 5.0 DH3-87:350-360 -20HP 117.0 671.0 635.0 42.0 DH3-87:360-370 -20HP 118.0 809.0 358.0 71.0 DH3-87:370-380 -20HP 261.0 1037.0 306.0 249.0 DH3-87:380-390 -20HP 1720.0 880.0 1432.0 54.0 DH3-87:390-400 -20HP 322.0 324.0 419.0 105.0 W-85-2:540-550 -20HP 101.0 1985.0 309.0 10.0 W-85-2:550-560 -20HP 121.0 1794.0 55.0 23.0 W-85-2:560-570 -20HP 198.0 288.0 324.0 395.0 W-85-2:580-590 -20HP 231.0 1697.0 415.0 23.0 W-85-2:580-590 -20HP 244.0 9108.0 1011.0 27.0 DH1-87:990-900 -20HP 53.0 710.0 255.0 18.0 DH1-67:900-910 -20 | | | | | 27 | . 2 • 0 |
| DHI-87:930-940 -60HN 625.0 535.0 117.0 5.0 DH3-87:350-360 -20HP 117.0 671.0 635.0 42.0 DH3-87:360-370 -20HP 118.0 809.0 358.0 71.0 DH3-87:370-380 -20HP 261.0 1037.0 306.0 249.0 DH3-87:380-390 -20HP 1720.0 880.0 1432.0 54.0 DH3-87:390-400 -20HP 322.0 324.0 419.0 105.0 W-85-2:540-550 -20HP 101.0 1985.0 309.0 10.0 W-85-2:550-560 +20HP 121.0 1794.0 55.0 23.0 W-85-2:560-570 +20HP 198.0 288.0 324.0 395.0 W-85-2:580-590 +20HP 231.0 1697.0 415.0 23.0 W-85-2:580-590 +20HP 244.0 9108.0 1011.0 27.0 DHI-87:990-900 +20HP 53.0 710.0 255.0 18.0 DHI-67:900-910 +20HP 53.0 205.0 151.0 <1.0 | DHI-87:920-930 | -GOHN | 270.0 | 243.0 | 168.0 | 4.0 |
| DH3-87:350-360 | DHI-87:930-940 | -GOHN | 625.0 | 535.0 | 117.0 | 5.0 |
| DH3-87:360-370 | DH3-87:350-360 | -20HP | 117.0 | 671.0 | 635.0 | |
| DH3-87:370-380 -20HP 261.0 1037.0 306.0 249.0 DH3-87:380-390 -20HP 1720.0 880.0 1432.0 54.0 DH3-87:390-400 -20HP 322.0 324.0 419.0 105.0 LW-85-2:540-550 -20HP 101.0 1985.0 309.0 10.0 W-85-2:550-560 -20HP 121.0 1794.0 55.0 23.0 W-85-2:560-570 -20HP 198.0 288.0 324.0 395.0 W-85-2:580-590 -20HP 231.0 1697.0 415.0 23.0 W-85-2:580-590 -20HP 244.0 9108.0 1011.0 27.0 DHI-87:390-900 -20HP 93.0 710.0 255.0 18.0 DHI-67:900-910 -20HP 53.0 205.0 151.0 <1.0 | DH3-87:360-370 | -20HP | 118.0 | 809.0 | | 71.0 |
| BH3-87:390-400 -20HP 322.0 324.0 419.0 105.0 W-85-2:540-550 -20HP 101.0 1985.0 309.0 10.0 W-85-2:550-560 -20HP 121.0 1794.0 55.0 23.0 W-85-2:560-570 -20HP 198.0 288.0 324.0 395.0 W-85-2:570-580 -20HP 231.0 1697.0 415.0 23.0 W-85-2:580-590 -20HP 244.0 9108.0 1011.0 27.0 BHI-87:890-900 -20HP 93.0 710.0 255.0 18.0 BHI-67:900-910 -20HP 53.0 205.0 151.0 <1.0 | DH3-87:370-380 | -20HP | 261.0 | 1037.0 | 306.0 | · · · · · · · · · · · · · · · · · · · |
| DH3-87:390-400 | .! - №5-07+30A-30A | 20Up | 1000 0 | 000 0 | 1400 0 | E: A A |
| W-85-2:540-550 | | | | | | |
| W-85-2:550-560 +20HP 121.0 1794.0 55.0 23.0 W-85-2:560-570 -20HP 198.0 288.0 324.0 395.0 W-85-2:570-580 +20HP 231.0 1697.0 415.0 23.0 W-85-2:580-590 -20HP 244.0 9108.0 1011.0 27.0 EHI-87:990-900 -20HP 93.0 710.0 255.0 18.0 EHI-67:900-910 -20HP 53.0 205.0 151.0 <1.0 | | | | | | |
| W-85-2:560-570 -20HP 198.0 288.0 324.0 395.0 W-85-2:570-580 -20HP 231.0 1697.0 415.0 23.0 W-85-2:580-590 -20HP 244.0 9108.0 1011.0 27.0 BHI-87:990-900 -20HP 93.0 710.0 255.0 18.0 BHI-67:900-910 -20HP 53.0 205.0 151.0 <1.0 | | | | | | |
| W-85-2:570-580 | | | | | | |
| W-85-2:580-590 -20HP 244.0 9108.0 1011.0 27.0 BHI-87:890-900 -20HP 93.0 710.0 255.0 18.0 BHI-67:900-910 -20HP 53.0 205.0 151.0 <1.0 | 1 | 2 VIII | 120.0 | 20010 | 024.0 | 374.0 |
| W-85-2:580-590 -20HP 244.0 9108.0 1011.0 27.0 BHI-87:890-900 -20HP 93.0 710.0 255.0 18.0 BHI-67:900-910 -20HP 53.0 205.0 151.0 <1.0 | W-85-2:570-580 | -20HP | 231.0 | 1697.0 | 415.0 | 23.0 |
| DHI-87:890-900 -20HP 93.0 710.0 255.0 18.0 DHI-87:900-910 -20HP 53.0 205.0 151.0 <1.0 | | | | | | |
| DHI-67:900-910 -20HP 53.0 205.0 151.0 <1.0 | | | | | · · | |
| | | | | | | |
| | BHI-87:910-920 | -20HP | 69.0 | 184.0 | 127.0 | 32.0 |

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AUTHORITY: C. FIPKE

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GEOCHEMICAL LABORATORY REPORT

| <i>-</i> - | SAMPLE TYPE | | | | | | | . (| CO | NC | EN] | RATE | | | |
|------------|-------------|---|----|---|---|------------|---|-----|----|------------|-----|--------------|----------------|----------------|---------------|
| S | A | М | F' | L | E | N | U | М | В | E | R | CU PPM | PB PPM | ZN PPM | MO PPM |
| | | | | | |)30)40 | | | - | 201 201 | | 95.0 94.0 | 217.0 283.0 | 139.0 119.0 | 81.0 119.0 |

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WORK ORDER: 51180-88

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GEOCHEMICAL LABORATORY REPORT

TOPE TECHNIQUES AND INSTRUMENTATION FOR THE EARTH SCIENCES

SAMPLE TYPE: CONCENTRATE

AG

SAMPLE NUMBER PPM

DHI-87:920-930 -20HP DHI-87:930-940 -20HP <0.2

0 -20HP <0.2

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GEOCHEMICAL LABORATORY REPORT

| | | SAM | P | L. | E | T | (F | E | : | C | 40 | 4C | E | NIRATE | · - · · · | • •••• | • | |
|-----------|------------|-------|-----------|-----|------------|------|------------|----|---|---|----------------|-----|-----|--------|------------------|------------|-------|---|
| ! | | | | | | | | | | | | | | CU | PB | ZN | MO | |
| S | Ê | H A | F | ? } | L | E | N | U | М | Ì | B | E | ĸ | PPM | PPM | PPM | PPM | |
| | DE | 1283 | 7: | 11 | 38 | 30-1 | 19 | 00 | | | -6 | 01 | łP | 40.0 | 348.0 | 178.0 | <1.0 | |
| : | Di | 1283 | 7 : | 1 |)(| 00-3 | 19 | 20 | | | -6 | ٥ŀ | O | 37.0 | 302.0 | 140.0 | <1.0 | |
| ! | Tel | 128 | 7 : | 1 | 92 | 20-1 | 19 | 40 | | | ۰- G | ٥ŀ | 4F | 43.0 | 535.0 | 227.0 | 2.0 | |
| i | DE | 1281 | 7: | 1 | 94 | 40- | 19 | 60 | | | -6 | ٥ŀ | łP | 93.0 | 453.0 | 182.0 | 3.0 | |
| | Di | 1281 | 7 | 1. | 9 6 | 50-3 | 19 | 80 | | • | G | ٥ŀ | łP | 325.0 | 280.0 | 250.0 | 4.0 | |
| 1 | DI- | 128 | 7: | 1 | 3 6 | 30-3 | 19 | 90 | | , | -6 | ٥ŀ | łF | 92.0 | 268.0 | 216.0 | <1.0 | • |
| | DI | 128 | 7 : | 1 | 38 | 30-1 | 19 | 00 | | | -6 | ٥ŀ | N | 220.0 | 650.0 | 564.0 | <1.0 | |
| | ITH | 128 | 7: | 1 | 9(| 00- | 19 | 20 | | | -6 | 01 | IN | 247.0 | 363.0 | 297.0 | 12.0 | 1 |
| ŀ | DF | 128 | 7 : | 1 | 9: | 20- | 19 | 40 | | | -6 | 01 | IN | 203.0 | 560.0 | 603.0 | 2.0 | |
| | DH | 128: | 1: | 1 | 9 4 | 40-1 | 19 | 60 | | • | - 6 | 01 | łИ | 154.0 | 490.0 | 508.0 | 2.0 | |
| | ĽΗ | (28) | 7: | 19 | 96 | 50-1 | เจ | 80 | | | -6 | 0 F | ١N | 1403.0 | 358.0 | 109.0 | 3.0 | |
| 1 | _ | | _ | | | 30- | | | | | | | IN | 255.0 | 367.0 | 254.0 | <1.0 | |
| | | | | | |)-9(| | | | | -6 | 01 | IN | 94.0 | 85.0 | 113.0 | <1.0 | |
| | | | | | |)-9: | | | | | -6 | Q٢ | ١N | | 110.0 | 141.0 | <1.0 | |
| I | H J | [-8] | 7 : | 9 | 1 (|)-9: | 2 O | | | • | -6 | 01 | IN | 48.0 | 80.0 | 254.0 | <1.0 | |
| D | н 1 | (-8: | 7: | 9: | 2 (|)-9: | 30 | | | | -6 | 01 | IN | 270.0 | 243.0 | 168.0 | 4.0 | |
| I | H] | [-8] | 7 : | 9: | 3 (| 9-9 | 40 | | | | - G | 01 | IN | 625.0 | 535.0 | 117.0 | 5.0 | |
| p | НЗ | 8-83 | 7: | 3: | 5(|)-3(| 50 | | | • | -2 | 01 | IF | 117.0 | 671.0 | 635.0 | 42.0 | |
| L | нэ | 8-8 | 7: | 3 | 5(|)-3: | 70 | | | , | -2 | 01 | (P | 118.0 | 809.0 | 358.0 | 71.0 | |
| I | НЗ | 3-87 | 7: | 3 | 7 (|)-3 | 30 | | | | -2 | 01 | łP | 261.0 | 1037.0 | 306.0 | 249.0 | |
| ! : I) | нз | 3-8: | 7: | 31 | 8 (|)-3° | 9 0 | | | | -2 | 01 | 1P | 1720.0 | 880.0 | 1432.0 | 54.0 | |
| | | | | | |)-4 | | | | | -2 | 01 | 42 | 322.0 | 324.0 | 419.0 | 105.0 | |
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AUTHORITY: C. FIPKE

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*** FINAL REPORT ***

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CERTIFICATE

- I, Peter Klewchuk, do hereby certify:
- 1. THAT I am a consulting geologist with office at 246 Moyie Street, Kimberley, B.C.
- 2. THAT I graduated in geology from the University of British Columbia in 1969 with a Batchelor of Science degree and from the University of Calgary in 1972 with a Master of Science degree.
- 3. THAT I am a Fellow of the Geological Association of Canada.
- 4. THAT I have practiced my profession for the past 16 years, primarily in western Canada and predominantly in southeastern British Columbia where I have become familiar with the Precambrian lithologies that underlie the Paul Mike claim group which is the subject of this report.
- 5. THAT I have no interest direct or indirect in the mineral claims herein reported, nor do I hold securities in any form, direct or indirect, in Dia Met Minerals Ltd.
- 6. THAT this report dated May 9, 1988 is based on personal involvement with the recent drilling on the property as well as reviews of published and unpublished reports of the claims and general area.
- 7. THAT I consent to the use of this report by Dia Met Minerals Ltd. as a statement of material facts.

Dated at Kimberley, British Columbia this 9th day of May, 1988.

Peter Kleiche

